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Survey Methodology

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Contemporary theory and practice of survey sampling: A celebration of research contributions of J.N.K. Rao

J.N.K. Rao is a Distinguished Research Professor in the School of Mathematics and Statistics at Carleton University, Canada. He is the world's leading researcher in the area of survey methodology and has profoundly influenced the field of sample surveys as used by government agencies and other organizations and businesses. Professor Rao received an MA from Bombay University in 1956 and a Ph.D. from Iowa State University in 1961. For more than 50 years, he has been a driving force in the development of unequal probability sampling methods, small sample approximations, analysis of complex survey data, empirical likelihood based inferences, variance estimation techniques and re-sampling methods, and missing data solutions with sound design-based properties. His abiding effort in meeting real world needs led to another prolific area of his research on small area estimation, highlighted by his book *Small Area Estimation* (1st edition in 2003 and 2nd edition with Molina in 2015) published by Wiley.

In addition to his phenomenal research impact, Professor Rao has had a significant influence on official statistics agencies through his participation on advisory boards and panels, and his role as advisor and consultant. He has also inspired several generations of survey statisticians through his teaching, mentoring and research collaboration. In particular, he mentored many Chinese statisticians who have become top researchers in Chinese universities.

During his remarkable and continuing academic career, Professor Rao has been honored by an array of prestigious academic awards, including the Gold Medal of the Statistical Society of Canada (1993), the Annual Morris Hansen Lecture (1998), the Waksberg Award (2005), the inaugural SAE Award (2017), and Honorary Doctorates from University of Waterloo, Canada (2008) and Catholic University of Sacred Heart, Italy (2013). He is Fellow of the American Statistical Association (1964), the American Association for the Advancement of Science (1965), and the Institute of Mathematical Statistics (1972). He was elected Fellow of the Royal Society of Canada in 1991.

On the occasion of Professor Rao's 80th Birthday, the Big Data Institute and the School of Mathematics and Statistics at Yunnan University, China, hosted a conference (May 24-27, 2017) celebrating Professor Rao's research contributions. Professor Jiahua Chen, the Director of the Big Data Institute and a long-time research collaborator of Professor Rao, was the Chair of the Organizing Committee. The conference brought together a distinguished group of researchers from many countries and presented a world-class scientific program on contemporary theory and practice in survey sampling.

In honour of Professor Rao's contributions, The *International Statistical Review* and *Survey Methodology* have agreed to publish joint special issues of papers presented at the conference. The special issue of the *International Statistical Review* features 15 papers. The first paper is a specially invited submission from Professor Rao on "My Chancy Life as a Statistician", which provides a brief account with amazing anecdotes on

^{1.} Song Cai, Carleton University. E-mail: scai@math.carleton.ca; Mahmoud Torabi, University of Manitoba. E-mail: Mahmoud.Torabi@umanitoba.ca.

his personal and research journey from India first to the United States and then to Canada. This paper is also reproduced in the *Survey Methodology* special issue. The remaining 14 papers in the *International Statistical Review* special issue are from all plenary speakers at the conference, covering diverse topics that reflect the current state-of-the-art research development in survey sampling. The *Survey Methodology* special issue contains 8 papers which are a subset of the remaining papers which were presented at the conference.

The joint special issues would not be possible without the unconditional support of the Co-Editors-in-Chief of the *International Statistical Review*, Drs. Ray Chambers and Nalini Ravishanker and the Editor of *Survey Methodology*, Wesley Yung. We would also like to use this opportunity to thank the sponsors of the conference, the Canadian Statistical Sciences Institute (CANSSI), the International Association of Survey Statisticians (IASS) of the International Statistical Institute (ISI), the International Chinese Statistical Association (ICSA), the International India Statistical Association (IISA), the Statistical Society of Canada (SSC), and Yunnan University, for their support.

Jiahua Chen, Yunnan University and University of British Columbia Changbao Wu, University of Waterloo Guest co-editors for the *International Statistical Review* Special Issue

> Song Cai, Carleton University Mahmoud Torabi, University of Manitoba Guest co-editors for the *Survey Methodology* Special Issue

My chancy life as a Statistician

J.N.K. Rao¹

Abstract

In this short article, I will attempt to provide some highlights of my chancy life as a Statistician in chronological order spanning over sixty years, 1954 to present.

Key Words: Bootstrap; Empirical likelihood; Linear mixed models; Small area estimation; Unequal probability sampling.

1 Introduction

Professor Changbao Wu, Guest Editor for this joint special issue between *ISR* and *Survey Methodology*, invited me to write an article tracing my chancy life as a statistician over the past 60 years. The joint special issue consists of papers based on plenary talks presented at a conference held in Kunming, China, May 24-27, 2017. This conference was sponsored by the Research Institute of Big Data, Yunnan University, and the organizing committee was chaired by Professor Jiahua Chen. I wish to first thank Professor Chen for organizing this conference "Contemporary Theory and Practice of Survey Sampling", celebrating my 80th birthday. I also wish to thank Professor Ray Chambers, Guest co-editor of *ISR* and Wesley Yung, Editor of *Survey Methodology*, for proposing this joint special issue, and to all the speakers for their excellent presentations. In this short article, I will attempt to provide some highlights of my chancy life as a Statistician in chronological order covering the period 1954-1958 in India, 1959-68 in USA with a one year break in 1963 in India, 1968-69 again in India and finally in Canada since 1969.

2 Early life in India

In 1954 I obtained a B.A. degree in Mathematics with specialization in Astronomy. I studied at a local college in my hometown Eluru, affiliated to Andhra University in India. Soon after writing my final exams, I was wondering what to do next and went to see my favorite algebra teacher, C.D. Murthy, for advice. He told me that I should study Statistics. I knew nothing about Statistics at that time but my mind was made up and I applied to some universities, including Bombay University, for admission. Only a few universities in India offered Statistics those days, only seven years after India achieved independence from Britain. But I was refused admission despite my first class in B.A. because my grades were not high enough. Only one student from Andhra University was admitted to Bombay University in 1954 for the Master degree in Statistics and his overall grade in his B. Sc. was 495 out of 500!

I was very frustrated and was wondering what to do next. My uncle, who studied in Bombay, advised me to go to Bombay and join the M.A. degree program in Pure and Applied Mathematics and try my luck afterwards. I was able to get admission and started my studies in Bombay three weeks later. But my mind

was set on Statistics and I did not enjoy the program I enrolled in except for the one course in Statistics I was allowed to take from the Statistics Department. Every week I used to see the Head of the Statistics Department, Professor M.C. Chakrabarti, to express my keen interest in pursuing a degree in Statistics. A month or so passed by and one fine morning, when I was at the Department to attend my statistics class, Professor Chakrabarti asked me if I would like to join his Department because one of the students left the program to study engineering in England. He also warned me that it would be extremely difficult for me to secure even a second class because I had no background in statistics and would be joining almost two months late. I took the chance and joined the program knowing that next year my chances will be slim again. First year was daunting and I managed to scrape through the first year unofficial examinations securing 23rd rank out of 24! I studied very hard next year and my enthusiasm for Statistics helped me a lot in my efforts. To my great surprise, I secured a First Class in the Final Examinations in 1956. (Only four students out of 24 secured first class that year if I remember correctly and that was a record compared to previous years!). I had great teachers including Chakrabarti and Anant Kshirasagar. I learnt a lot from them even though some of the stuff was boring (like working out the recurrence relations for the moments from Kendall's book!). Chakrabarti taught sampling theory and I got attracted to it. Also, it was fortunate for me that three classic books in survey sampling by Cochran, Sukhatme and Hansen, Hurwitz and Madow appeared around 1954. I might add that India produced some great statisticians by that time, including C.R. Rao, R.C. Bose, P.C. Mahalanobis and P.V. Sukhatme. Indian statisticians owe much to Mahalanobis for his vision and pioneering contributions in promoting Statistics in India and putting India on the world map.

After finishing my M.A. degree, I wanted to take a job so that I could support my family (my father died when I was only six), but my mother insisted that I should pursue a Ph.D. degree. Chakrabarti offered me a Government of India Senior Research Scholarship to work with him on the construction of experimental designs but I was not strong in that subject and also had no interest. I applied to the Indian Statistical Institute for a research scholarship without success but I was admitted to the second year of a three- year Diploma course. I joined that program but most of the stuff was a repeat of what I learnt in my M.A. program. At that time Dr. K.R. Nair, well-known for his work on the construction of experimental designs, was looking for a research scholar to work with him at the Forest Research Institute in Dehra Dun, India. I joined him in October 1956 as a research scholar. Seeing my interest in survey sampling, he encouraged me to work on problems related to forest surveys. He also felt that I should go abroad to do my Ph.D. I managed to publish few papers on sampling related to forest surveys. At that time Professor H.O. Hartley was doing great work at Iowa State University (ISU) on survey sampling. Nair studied with Hartley in London, so he advised me to apply to ISU to work with Hartley. Again I was not admitted right away but a chance vacancy occurred and I ended up in Ames, Iowa around the middle of the fall quarter of 1958.

3 Life in USA: 1958-68

Undoubtedly, ISU was among the best (if not the best) applied statistics departments at that time. (I believe it still is.) I even had the chance to take the last course on statistical methods with George Snedecor

before he retired. He was the founder of the Statistics Department at ISU and his close association with R.A. Fisher led to the well-known Snedecor's F and also Fisher going to ISU as a visiting professor. It was most rewarding to learn from great statisticians like Hartley and Kempthorne at ISU and also from others who visited ISU regularly. Professor Hartley was my mentor and Ph.D. supervisor and I learnt from him that the development of statistical theory should be motivated by practical applications. I took economics as a minor in my Ph.D. program and I was fortunate to work with Gerhard Tintner who was a pioneer in Econometrics and one of the inventors of the Variate Difference Method for finding the order of difference that makes a time series stationary. I even wrote two papers and a small monograph with him on this topic. For several years I tried to keep up with the developments in Econometrics.

I stayed at ISU for 5 years, three years as a student and two years as Assistant Professor, before returning to India in 1963 for family reasons. This period was most exciting and professionally rewarding. At that time unequal probability sampling without replacement was a "hot" topic and people were looking for practical procedures. Hartley and I published a paper on this topic in the Annals of Statistics (1962) developing an asymptotic theory for randomized probability proportional to size (PPS) systematic sampling (Hartley and Rao, 1962) After finishing my Ph.D. in 1961, I published a paper with Hartley and W.G. Cochran, in the Journal of the Royal Statistical Society, Series B, 1962, on a very simple procedure of unequal probability sampling without replacement that has many desirable properties (Rao, Hartley and Cochran, 1962). This method is now known as the RHC method and many papers on this method have appeared since then. Both the PPS systematic sampling method and the RHC method have been used in the Canadian Labour Force Survey for the past 25 years or so. Professor Arijit Chaudhuri of the Indian Statistical Institute has used the RHC method extensively for large-scale sample surveys in India. I also wrote a paper in the Journal of the American Statistical Association on composite estimation for repeated surveys with my Canadian friend, Jack Graham, who was also a student at ISU at that time (Rao and Graham, 1964). Jack became my colleague after I joined Carleton University in Ottawa in 1973. More recently, I got back to composite estimation in the context of the Canadian Labour Force Survey and developed a new method in association with Wayne Fuller and Avi Singh, that is currently being used in Canada (Fuller and Rao, 2001 and Singh, Kennedy and Wu, 2001). I shared an office with Wayne Fuller at ISU and he has been a close friend for the past 55 years.

I worked as a sample survey expert at the National Council of Applied Economic Research in New Delhi for one year after I returned to India. During my stay there I was involved in the development of the design and analysis of an All India Consumer Expenditure Survey. But I was very frustrated because there were no facilities there for research. I returned to United States in August 1964 and worked for one year in Dallas in a research group headed by D.B. Owen before joining Hartley at Texas A&M University. (Hartley moved to Texas A&M in 1963 to create an Institute of Statistics there.) My stay at Texas A&M was also most rewarding and professionally exciting. I worked closely with Hartley and also supervised Ph.D. students. I was promoted to Full Professor rank in 1967 and things were going great. My son, Sunil, was born in April 1967 and we were well settled. But I had to return to India in June 1968 due to unexpected family problems.

I took leave from Texas A&M and joined the Indian Statistical Institute (ISI) in Calcutta as Visiting Professor. (I might mention here that my son Sunil is currently Director of Biostatistics Division and Interim Chair of the Department of Public Health Sciences at the University of Miami. He was elected ASA Fellow in 2011 and we two belong to the very small group of father-son ASA Fellows!)

I would like to briefly mention four significant contributions I made during my stay at Texas A&M. In my Biometrika 1967 paper with Hartley, we gave a matrix formulation of general ANOVA mixed models that was instrumental to the derivation of maximum likelihood (ML) estimators of both fixed effects and variance components (Hartley and Rao, 1967), We also developed an EM algorithm in this paper but did not pursue it further due to computational limitations at that time. (EM algorithm became popular after the appearance of Dempster, Laird and Rubin (1977)). Patterson and Thompson (1971) modified our ML method and developed restricted maximum likelihood (REML) estimation. Many extensions and refinements have been made over the past 40 years, and several software packages implemented those methods. An excellent review paper by Harville (1977) contributed to the extensive use of those methods. I also worked with Hartley on variance estimation when only one unit is sampled from each stratum (Hartley, Rao and Kiefer, 1969). In this case, standard design-based methods are not applicable and it is necessary to resort to models. We used a linear regression model with unequal error variances and expressed the variance of the stratified mean as a linear combination of the error variances. We then developed a new method of estimating the error variances that in turn led to a new variance estimator for the stratified mean. We submitted this paper for publication in 1968 before I left for India. I gave a seminar talk at ISI on this work. After my talk, Professor C.R. Rao felt that he could establish some optimality properties for our method. This led to C.R. Rao's well-known MINQUE method (Rao, 1970), and Professor Rao notes "The motivation for writing this article is a recent contribution by Hartley, Rao and Kiefer (1969) who obtained unbiased estimator when all the variances are unequal ..." (page 161).

In the 1960's, V.P. Godambe was giving talks at various professional meetings on his important contributions to survey sampling inference; in particular, on the non-existence of a best estimator in a general class of linear unbiased estimator of a total and on the flat likelihood caused by the label property of a finite population. Those negative results are indeed fundamental, but Hartley and I felt that some of the alternative criteria proposed for the choice of an estimator, such as admissibility and hyper-admissibility for any sampling design, are unsatisfactory. In our Biometrika 1968 paper we suggested that some aspects of the sample data, depending on the situation at hand, need to be ignored to arrive at an informative likelihood (Hartley and Rao, 1968). We proposed such a non-parametric likelihood that is now called Empirical Likelihood (Owen, 1988). We also showed how to incorporate known population totals of auxiliary variables, and showed that the empirical likelihood (EL) estimator of a total is close to a regression estimator. I gave several lectures on the foundations of inference in survey sampling at ISI and Professor C.R. Rao wrote a nice article afterwards (Rao, 1970) that seems to agree with our approach: "In situations like the one we are considering where the full likelihood does not satisfy our purpose, we may have to depend on a statistic which for every observed value supplies information (however poor it may be) on

parameters of interest." Our Biometrika 1968 paper also contained a short section on Bayesian inference for the mean obtained by combining our likelihood with a diffuse conjugate prior. Ericson (1969) combined Godambe's flat likelihood with an informative prior to produce informative posterior inferences on the mean. Our results are algebraically identical to Ericson's, but fundamentally different in the sense that our inferences depend on the probability distribution induced by the survey design, unlike Ericson's results.

While I was working on my Ph.D. thesis at ISI, I analyzed some farm survey data where the farms were selected with probabilities proportional to farm sizes. I found that some variables of interest, in particular poultry size, was unrelated to farm size and that the use of the widely used Horvitz-Thompson (HT) unbiased estimator in such cases would lead to very large variances. I therefore proposed an alternative estimator that ignores the survey weights but uses the population structure (Rao, 1966). I provided both theoretical and empirical justifications for preferring such an estimator. My result essentially casts doubt on the usefulness of criteria that advocate the HT estimator for ANY design and ANY characteristic. Later, D. Basu used an amusing circus elephant example to demonstrate that the HT estimator leads to absurd results if the sizes are unrelated to the values of interest (Basu, 1971).

4 Life in Canada: 1959-2000

I found that the Canadian universities suited my family circumstances in India at that time and decided to migrate to Canada in 1969 directly from Calcutta. Hartley was very unhappy with my decision, but we continued our collaboration for several years. I worked four years at the University of Manitoba before joining Carleton University, Ottawa in 1969. I have also worked at Statistics Canada for the past 40 years or so as a consultant, and this practical exposure was extremely useful in my later research work. I have collaborated with many statisticians over the past 25 years, thanks to my Canadian NSERC research grant that encourages collaborative work. I supervised many outstanding Ph.D. students in Canada. My first Ph.D. student in Canada, David Bellhouse (co-supervised with Jim Kalbfleish at the University of Waterloo), wrote his thesis on optimal estimation in finite population sampling. He had a distinguished career at the University of Western Ontario and retired recently. Bellhouse is also a leading expert in the history of Statistics. Dan Krewski was my first Ph.D. student at Carleton University. He developed asymptotic theory for stratified multistage sampling designs (Krewski and Rao, 1981) which provided theoretical justification for replication methods, such as the jackknife and balanced repeated replication, widely used for the analysis of complex survey data (see Shao and Tu, 1995, Chapter 6). Krewski is currently a distinguished professor of biostatistics and population health at the University of Ottawa and he is a leading authority on risk assessment. Both Bellhouse and Krewski are ASA Fellows. Several of my Masters and Ph.D. students established successful careers at Statistics Canada and elsewhere.

In 1977, I was looking for a suitable place to spend my sabbatical leave. By chance, I bumped into Fred Smith of the University of Southampton at a survey sampling conference held at the University of North Carolina. He mentioned that he has applied for a research project on the analysis of complex survey data

and if successful I could spend my sabbatical leave at his university working on the project. His research project was approved and I joined the project team (Fred Smith, Tim Holt, Gad Nathan and Alastair Scott) in April 1978 for 4 months. I also had a chance to interact with Graham Kalton who was also at the University of Southampton. I might mention that Smith, Holt and Kalton developed a strong program in survey sampling research at the University of Southampton. In later years, Chris Skinner, Ray Chambers and Danny Pfeffermann contributed greatly and made it into a leading center for survey research.

During my sabbatical leave, Alastair Scott and I worked on methods for the analysis of categorical survey data and published several papers subsequently. In Rao and Scott (1981, 1984), we developed simple corrections to standard chi-squared tests for testing independence in a two-way table of weighted counts that account for the survey design features. It was nice to see the 1981 paper with Scott included among the 19 landmark papers in survey sampling published over the period 1930-90. Scott visited me regularly for several years to continue our work on analysis of survey data and other topics until his health did not permit him to travel alone. He was suffering from brain cancer but hoped to attend the China conference in May 2017. I was deeply saddened by the news of his death on the first day of the conference. I would like to dedicate this joint special issue of *ISR* and *Survey Methodology* to the memory of my dear friend and collaborator, Alastair Scott.

I collaborated with several excellent researchers after my return from sabbatical leave. Jeff Wu and I developed valid bootstrap variance estimators for stratified multistage sampling and other designs (Rao and Wu, 1988) and we introduced the concept of bootstrap weights (Rao, Wu and Yue, 1992). Currently, bootstrap weights are used at Statistics Canada for variance estimation in several large-scale surveys. Other major collaborations include the following: (1) multiple frame surveys with Chris Skinner, Sharon Lohr and Changbao Wu, (2) empirical likelihood intervals for survey data with Changbao Wu, Jiahua Chen, Yves Berger and M. Salehi, (3) analysis of survey data with Alastair Scott, Chris Skinner, Roland Thomas, Mike Hidiroglou, Wesley Yung and Jun Shao, (4) imputation for missing data with Jun Shao, Randy Sitter, Jae Kim, Qihua Wang, Jiahua Chen and Y.S. Qin. Randy Sitter and Jun Shao were my colleagues during the period 1990-95, and our statistics group was rated among the top 15 in the world for research productivity. Other collaborators include Arun Nigam, Jurgen Kleffe, K. Vijayan, Avi Singh, Gordon Brackstone, Poduri Rao and P.A.V.B. Swamy.

Around 1985, I got interested in small area estimation after organizing an international symposium on small area statistics in 1985 jointly with Statistics Canada. Invited papers presented at the symposium are published in a Wiley book (Platek, Rao, Särndal and Singh, 1987). Demand for reliable small area statistics has steadily grown in the past 25 years which in turn led to many theoretical and practical contributions. I supervised several Ph.D. students on this topic, including N.G.N. Prasad, Diane Stukel, Ming Yu and Yong You. Prasad developed accurate mean squared error estimators of model-based small area estimators (Prasad and Rao, 1990) and this work is widely cited. Yong You received the Pierre Robillard award of the Statistical Society of Canada for the best Ph.D. thesis in the year he graduated.

5 Post retirement: 2000-present

I took early retirement in 2000, two years before the mandatory 65, but I have not really slowed down since my retirement 17 years ago. I almost died in 2002 of cardiac arrest without any prior symptoms, but by chance it happened in the hospital and I was saved. I was able to complete my Wiley book on small area estimation (Rao, 2003) and I am happy to see that it is well received and highly cited. I had excellent collaborators in small area estimation (SAE), including Isabel Molina, Malay Ghosh, Partha Lahiri, Gauri Datta, Jiming Jiang, Bal Nandram, Kalyan Das, Sharon Lohr, Domingo Morales, Leyla Mohadjer, Hussain Chowdhry and Tatsuya Kubokawa. By chance, I met Isabel Molina at the ISI meetings in Lisbon and she invited me to Madrid to give a workshop. This led to close collaboration on SAE with her and our paper on empirical Bayes (EB) estimation of complex small area parameters, such as poverty indicators, received the best paper award in 2010 from the Canadian Journal of Statistics (Molina and Rao, 2010). Measurement of poverty indicators for small areas received considerable attention after the World Bank promoted a method based on simulated censuses. In the 2010 paper we showed that the EB method can be considerably more efficient. I also collaborated with Molina on the second edition of my Wiley book (Rao and Molina, 2015). I was very fortunate to have two excellent students, M. Torabi and M. Diallo, working on SAE after my retirement. I also supervised another excellent student, David Haziza, on missing data and imputation. All three are "rising stars" and Haziza is also an ASA Fellow and received the prestigious Gertrude Cox Award for 2018.

I am happy that several of my collaborators participated in the China Conference as plenary speakers and contributed to this joint special issue of *ISR* and *Survey Methodology*. My thanks are due to them as well as to other speakers who have contributed to the joint special issue.

All in all, my chancy life as a Statistician has been very rewarding and satisfying. It was a great pleasure to work with many excellent researchers and graduate students. I owe it to my algebra teacher C. D. Murthy, to Professor M.C. Chakrabarti, to my mentor Professor H.O. Hartley, to my mother and to my wife for whatever success I have achieved in my chancy life as a Statistician over the past 60 years.

Acknowledgements

An earlier version of this paper appeared soon after my retirement in a 2004 newsletter of the International Indian Statistical Association (IISA). I thank the IISA Executive for giving permission to update the paper for publication in the joint special issues of *ISR* and *Survey Methodology*.

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Bayesian small area demography

Junni L. Zhang, John Bryant and Kirsten Nissen¹

Abstract

Demographers are facing increasing pressure to disaggregate their estimates and forecasts by characteristics such as region, ethnicity, and income. Traditional demographic methods were designed for large samples, and perform poorly with disaggregated data. Methods based on formal Bayesian statistical models offer better performance. We illustrate with examples from a long-term project to develop Bayesian approaches to demographic estimation and forecasting. In our first example, we estimate mortality rates disaggregated by age and sex for a small population. In our second example, we simultaneously estimate and forecast obesity prevalence disaggregated by age. We conclude by addressing two traditional objections to the use of Bayesian methods in statistical agencies.

Key Words: Small area estimation; Bayesian hierarchical model; Weakly informative prior; Life expectancy; Obesity; New Zealand; Forecasting.

1 Introduction

Demography has traditionally been a big-data and big-area discipline. Demographers have used censuses, registration data, and surveys to obtain national-level estimates and forecasts. Big sample sizes for national populations have meant that, in contrast to most of applied statistics, sampling variation is small. Demographers have instead concentrated on other problems, such as measurement errors, and developed their own techniques and terminology distinct from mainstream statistics. Classic demographic methods combine simple deterministic models with complex expert judgements. The models are simple enough to be implemented on computer spreadsheets, but require practitioners to intervene and correct for problems caused by violations of the underlying assumptions. These methods have had many successes. They have, for instance, been used to document the dramatic fall in mortality and fertility in developed countries, and have alerted policy makers to future population ageing.

Traditional demographic methods are, however, coming under strain. The reason is the rising demand for disaggregation. Policy makers, social scientists, market researchers, and other users of demographic estimates and forecasts require ever-more disaggregated numbers. The United Nations 2030 Agenda for Sustainable Development, for instance, calls for increasing significantly "the availability of high-quality, timely and reliable data disaggregated by income, gender, age, race, ethnicity, migratory status, disability, geographic location and other characteristics relevant in national contexts" (United Nations General Assembly, 2015, Goal 17.18). Disaggregation is challenging to traditional demography because, even when the overall population is large, the number of people in each subpopulation can be small. With these small numbers, random variation in data collection, or in underlying demographic processes such as fertility, mortality, and migration, becomes prominent, and deterministic methods break down.

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To deal with these problems, demographers have been turning to mainstream statistics for new ideas on ways to deal with random variation. Similarly, statisticians have been showing an increasing interest in demographic applications. The result has been a boom in statistical demography (Alho and Spencer, 2006).

Demographic phenomena are often highly regular. Mortality, fertility, and migration rates, for instance, have characteristic age-sex profiles that are stable over time or that change in consistent ways. These regularities reflect common events over individuals' life courses. Migration rates typically peak in the late teenage years, for instance, because these are the years when people reach adulthood and begin to leave home. The ability to model units that are similar but not identical is a particular strength of Bayesian methods. Bayesians build models with multiple layers that can capture multiple, overlapping types of variability. Bayesian models pool information from across similar units, to improve accuracy and precision.

Bayesian methods have other advantages for demographic modelling. They can coherently combine uncertainty from many sources, including random variation, missing data, and uncertainty about future trends. Bayesian methods also make it easy to construct inferences about derived quantities. Life expectancy, for instance, is a complicated nonlinear deterministic function of age-specific mortality rates, but within a Bayesian framework, deriving inferences about life expectancy from inference about age-specific mortality rates is straightforward.

Because of advantages such as these, within the field of statistical demography, there has been particularly fast growth in *Bayesian* statistical demography (Bijak and Bryant, 2016). The most prominent example has been the adoption, by the United Nations, of Bayesian methods for population forecasting (Gerland, Raftery, Ševčíková, Li, Gu, Spoorenberg, Alkema, Fosdick, Chunn, Lalic, Bay, Buettner, Heilig and Wilmoth, 2014).

In this paper, we illustrate how Bayesian methods, and particularly Bayesian hierarchical models, can be used to obtain disaggregated demographic estimates and forecasts. The examples are drawn from a longterm project to develop Bayesian demographic methods for use in official statistics, including the development of open source software implementing the methods. In the statistical literature, the problem of obtaining estimates for domains with small sample sizes has been referred to as small area estimation (Pfeffermann, 2013; Rao and Molina, 2015). The models that we consider are all "area-level" models, in that they use counts and rates for disaggregated cells, rather than individual-level data. With area-level models, we can use datasets in the form of confidentialized tables that individual-level models cannot use. Demands for disaggregated estimates and forecasts are also related to groups rather than individuals.

In Section 2, we present mortality estimates for Māori, the indigenous people of New Zealand. The main inferential challenge is to capture the complex relationship between mortality and age, despite small numbers and considerable random variation. In Section 3, we interpolate and forecast obesity rates in New Zealand by age, based on survey data. The main problem here is carrying out a time series analysis with data from only five years. We conclude, in Section 4, by addressing two traditional objections to the use of Bayesian methods in statistical agencies.

2 Mortality rates for Māori

2.1 The estimation problem

Mortality rates are a fundamental measure of human welfare, as well as a major performance indicator for the health sector. Mortality rates are also a key input for population forecasts, and for the life insurance industry.

New Zealand's national statistical office (Stats NZ), publishes estimates of mortality rates for Māori by sex and by one-year age groups. These rates are "super-population" estimates. Super-population mortality rates measure the underlying risk of dying. They can be contrasted with finite-population rates, which measure the actual number of deaths divided by the actual population at risk. Suppose, for instance, that no 6-year-old Māori die in a particular year. The finite-population mortality rate is exactly zero, but the underlying risk of dying, and hence the super-population mortality rate, is presumably non-zero.

To derive death rates we need death counts and measures of population at risk. New Zealand's death registration system is efficient and complete, and reporting of ethnicity on the death registrations is generally reliable (Bryant and Howard, 2017), so data on death counts can be treated as error-free. Finding appropriate measures of population at risk is more challenging. Population at risk is measured using person-years. For instance, if a person spends 9 months in New Zealand during the period of interest, then that person contributes 0.75 person-years to the population at risk. Ideally, population at risk would be obtained by summing up person-years contributed by each person in the population. However, such data can be difficult to obtain. Instead, demographers typically approximate population at risk using population count multiplied by length of period. Population counts for Māori in New Zealand are relatively accurate for census years (Bryant, Dunstan, Graham, Matheson-Dunning, Shrosbree and Speirs, 2016), but become less accurate away from census years, because it is not possible to tell, from international migration data, how many Māori are entering and leaving the country. In addition, Stats NZ does not treat ethnicity as a characteristic that is fixed at birth, but rather as an aspect of personal identity that individuals can change over their lifetimes.

In response to the difficulties in estimating Māori population counts outside census years, Stats NZ focuses on periods centered on census years. Censuses are normally carried out every 5 years in New Zealand, though the 2011 census was postponed until 2013 because of an earthquake. The standard approach to mortality estimation is to use three-year periods, centered on a census year, such as 2012-2014. Using a three-year period gives larger numbers of death counts in each age-sex cell, and hence more stable estimates, than would be the case with single-year periods. To approximate the population at risk over a three-year period, Stats NZ uses the population count at the middle of the period, that is on June 30 of the census year, multiplied by 3.

To give an idea of the modelling challenge, Figure 2.1 shows direct estimates of mortality rates on a log scale for Māori males in 2012-2014, for single-year age groups 0, 1, ..., 100+. Direct estimates of mortality rates are simply death counts for each age-sex cell divided by the population at risk for that cell. The diameter of each circle in Figure 2.1 is proportional to the square root of the number of deaths. Altogether,

there were 9,170 deaths during the period, with the largest cell consisting of 130 deaths, two cells having 0 deaths, and a median death count of 27. The Māori male population on June 30, 2013 was 328,000, giving a population at risk of $328,000 \times 3 = 984,000$ person-years.

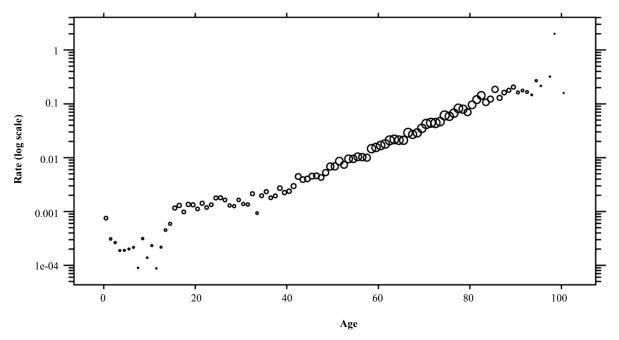


Figure 2.1 Direct estimates of mortality rates for Māori males in 2012-2014, by single year of age. The diameter of the circles are proportional to the square root of the number of deaths during the period.

Estimating underlying death rates between ages 40 and 90 is relatively easy. There are plenty of data, and, when shown on a log scale, the rates appear to fall on a straight line.

Somewhere between age 10 and age 20, death rates rise sharply, and then climb slowly up to about age 35. Many countries have a similar pattern of unusually high mortality rates in the late teenage years and 20s, particularly among males. The phenomenon is referred to as the "accident hump" (meaning, mainly, car accidents), though in many places, including New Zealand, it would be more accurate to call it an accident and suicide hump.

Death rates are relatively high during the first year of life, before falling to very low levels. Exactly how low these rates go is difficult to tell, because death counts are small and the associated direct estimates are highly erratic. The same problem also exists above age 90, where trends in death rates are difficult to pin down.

The death rate for 99-year-olds is over 1. This implies that the number of deaths of 99-year-olds is greater than the (approximate) number of person-years lived during the period 2012-2014 by 99-year-olds. Rates, unlike probabilities, have no upper bound. Consider, for instance, a population consisting of one person, who dies 9 months into a one-year period. The implied death rate for that period is $1/0.75 \approx 1.33$.

2.2 The model

2.2.1 Model specification

Our input data are death counts y_{ast} and population at risk n_{ast} . Subscript *a* denotes age group; subscript *s* denotes sex; and subscript *t* denotes time, taking values 2005-2007 and 2012-2014. Using two periods allows us to borrow strength across time, and also to study change over time.

We model death counts as draws from a Poisson distribution,

$$y_{ast} \sim \text{Poisson}(\gamma_{ast}n_{ast}),$$
 (2.1)

where γ_{ast} is the super-population mortality rate. We calculate n_{ast} by multiplying the population at June 30 in the census years by 3, and treat it as error-free. The main goal of the modelling is to estimate γ_{ast} .

Traditionally, demographers have ignored the fact that, even after knowing the population at risk and the underlying death rate, the actual number of deaths is still random and therefore uncertain. With large cell counts, such as for national populations, this uncertainty is small, so ignoring it is sensible. With small cell counts, however, this uncertainty is substantial, and needs to be accounted for. We do this by treating y_{ast} as a random draw from a Poisson distribution.

We add to (2.1) assumptions about how γ_{ast} is likely to vary. In Bayesian terminology, we specify a prior model for the γ_{ast} . Because γ_{ast} is positive with no upper bound, we specify the model on a log scale. We assume that γ_{ast} varies systematically by age, sex, and time, with age patterns potentially differing between females and males,

$$\log \gamma_{ast} = \beta^0 + \beta_a^{age} + \beta_s^{sex} + \beta_t^{time} + \beta_{as}^{age:sex} + e_{ast}.$$
 (2.2)

Here, β^0 is an intercept, capturing the overall level of log mortality rates, β_a^{age} is an age effect, capturing variation across age, β_s^{sex} is a sex effect, capturing variation between sexes, β_t^{time} is a time effect, capturing common time trends, and $\beta_{as}^{\text{age:sex}}$ is an age-sex interaction, capturing variation between sexes in the age pattern. The presence of the error term e_{ast} , implies that we do not expect our prior model to predict $\log \gamma_{ast}$ with complete accuracy. Standard generalized linear models do not have an equivalent term, and thus are implicitly making stronger assumptions about the correctness of the model. We assume that the error term e_{ast} has a normal distribution with mean 0 and variance σ^2 . The higher the value of σ^2 , the less the implied accuracy of the prior model.

The most importance source of variation in mortality rates is age. As is apparent in Figure 2.1, mortality rates for people in the 90s are three or four orders of magnitude higher than mortality rates for young children. It is therefore crucial for accurate estimation that we capture the main features of the age pattern.

We model age effects using an approach originally developed for modelling change over time rather than age, a "local trend" model (Prado and West, 2010, pages 119-121),

$$\beta_a^{\text{age}} = \alpha_a^{\text{age}} + 1(a=0)\psi + u_a^{\text{age}}, \qquad (2.3)$$

$$\alpha_a^{\text{age}} = \alpha_{a-1}^{\text{age}} + \delta_{a-1}^{\text{age}} + v_a^{\text{age}}, \qquad (2.4)$$

$$\delta_a^{\text{age}} = \delta_{a-1}^{\text{age}} + w_a^{\text{age}}. \tag{2.5}$$

Use of time series models to capture variation over age is relatively common in statistical demography. The fundamental idea is that values for neighboring age groups, like values for neighboring time periods, are more highly correlated than values for age groups or time periods that are distant from one another.

Equation (2.3) says that age effects are a combination of underlying level, captured by α_a^{age} , and agespecific idiosyncratic effects, captured by error term u_a^{age} . Age group 0 typically has much higher mortality rates than those for other young age groups, reflecting the special risks faced by infants. This extra mortality is modelled by parameter ψ . Equation (2.4) says that the level effect at age a equals the level effect at age a - 1, plus an increment $\delta_{a-1}^{\text{age}}$, plus an idiosyncratic error v_a^{age} . Equation (2.5) says that the increment at age a, δ_a^{age} , equals the increment at age a - 1, $\delta_{a-1}^{\text{age}}$, plus an idiosyncratic error w_a^{age} . Under a local trend model, age effects are expected to rise or fall linearly, but the slope of the line can change, or even reverse direction, over the whole length of the age pattern. Our priors for the starting values of α^{age} and δ^{age} are $\alpha_0^{\text{age}} \sim N(0, 10^2)$ and $\delta_0^{\text{age}} \sim N(0, 1)$.

The age-sex interaction term $\beta_{as}^{age:sex}$ measures variation between sexes in the age pattern for mortality. We use a "local level" model (Prado and West, 2010, pages 119-121),

$$\beta_{as}^{\text{age:sex}} = \alpha_{as}^{\text{age:sex}} + u_{as}^{\text{age:sex}}, \qquad (2.6)$$

$$\alpha_{as}^{\text{age:sex}} = \alpha_{a-1,s}^{\text{age:sex}} + v_{as}^{\text{age:sex}}, \qquad (2.7)$$

 $s \in \{\text{Female, Male}\}$. This model expresses the idea that, after accounting for age effects and sex effects, the residuals for mortality rates will be similar between neighbouring age groups, within each sex. The lack of a trend term (δ) implies that we do not expect these residuals to systematically trend upwards or downwards across the age range. We assume that any systematic trend will be shared by both sexes, and hence will be accounted for by the trend term in the age effect. Our prior for the starting value of $\alpha^{\text{age:sex}}$ is $\alpha_0^{\text{age:sex}} \sim N(0, 10^2)$.

We use a simple model for sex effects,

$$\beta_s^{\text{sex}} \sim \mathcal{N}(0, 1), \qquad (2.8)$$

 $s \in \{\text{Female, Male}\}$. This implies that we expect that the mean difference between mortality rates for sex s and the average mortality rates for both sexes to lie within the range (-2, 2) on a log scale. The variance of the female-male differences is 1 + 1 = 2, so we expect this difference to lie within the range $(-2\sqrt{2}, 2\sqrt{2})$ on a log scale, or (0.06, 16.9) on the original scale, which is a very large range compared to actual sex differences. This is an example of a "weakly informative" prior, in that it understates the actual strength of existing scientific knowledge (Gelman, Jakulin, Pittau and Su, 2008). Weakly informative priors provide many of the benefits of strong priors, by ruling out implausible values, and speeding up

computations. However, they are much more convenient, since they do not require the analyst to precisely summarize external information about the parameter in question, which can be difficult.

As there are only two time periods, and hence insufficient information to warrant a complicated model for time effects, we simply assume that $\beta_t^{\text{time}} \sim N(0, 1)$. We assume $\beta^0 \sim N(0, 10^2)$.

All the error terms in our model $(e_{ast}, u_a^{age}, v_a^{age}, w_a^{age}, u_{as}^{age:sex}, and v_{as}^{age:sex})$ have normal distributions with mean 0. The standard deviation parameters for the error terms e_{ast} , u_a^{age} , v_a^{age} and w_a^{age} all have a half-t distribution, with 7 degrees of freedom and scale parameter 1. Figure 2.2 shows a half-t distribution with 7 degrees of freedom and scale parameter 1. The distribution puts a 65% probability on values below 1, and a 2% probability on values exceeding 3.

In practice, we expect the standard deviation of our error terms to be well under 1. The standard deviation governs the size of age-to-age, sex-to-sex, or time-to-time differences in rates. A standard deviation of 1 implies that we would often see differences of 100% or more, which we do not see in practice. Our prior for standard deviations is therefore weakly informative.

The standard deviation parameters for the error terms $u_{as}^{age:sex}$ and $v_{as}^{age:sex}$ in the age-sex interaction have a half-t distribution, with 7 degrees of freedom and scale parameter 0.5. We use a smaller scale for the interaction on the principle that interactions are typically smaller in size than main effects (Gelman et al., 2008).

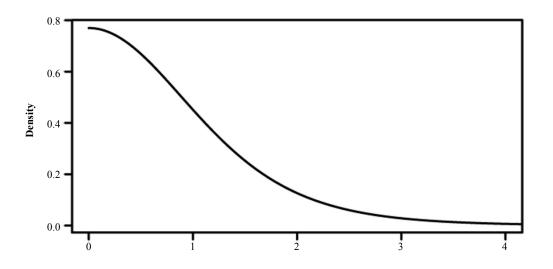


Figure 2.2 A half -t distribution with 7 degrees of freedom and scale 1.

2.2.2 Model output

The output from a Bayesian analysis is a sample from the posterior distribution of all of the parameters conditional on the data. The parameters include mortality rates, disaggregated by age, sex, and time,

parameters in the model for age effects, parameters in the model for sex effects, and so on. Our main interest lies in the mortality rates. The mortality rates are well identified from the data. The main effects and interactions in the prior model, in contrast, are only weakly identified. We discuss the identification issue further in Appendix A.

We simulate draws from the posterior distribution using 4 independent chains, each with a burnin of 100,000 and production of 100,000. We keep every 250^{th} iteration from each chain, yielding a combined sample of S = 1,600 draws. We monitor convergence using potential scale reduction factors (Gelman, Carlin, Stern, Dunson, Vehtari and Rubin, 2014, Section 11.4). The calculations are done in our own open source R package *demest*. Sample code is shown in Appendix B.

For any given parameter, we use the median of its posterior draws as the point estimate, and use the 100p% (0) credible interval formed by the <math>[50(1-p)] and [50+50p] percentiles of its posterior draws to measure the uncertainty. For instance, a 95% credible interval with p = 0.95 is formed by the 2.5% and 97.5% percentiles of the posterior draws.

The posterior draws can easily be used to construct estimates of functions of the model parameters, together with measures of uncertainty. In the study of mortality, a particularly important example is life expectancy at birth. Life expectancy is a complicated nonlinear function of age-specific mortality rates (Preston, Heuveline and Guillot, 2001). Let $f(\gamma)$ denote the nonlinear function that produces life expectancy from a set of age-specific mortality rates γ . If $\gamma^{(1)}, \ldots, \gamma^{(S)}$ represent a sample from the posterior distribution of γ , then $f(\gamma^{(1)}), \ldots, f(\gamma^{(S)})$ form a sample from the posterior distribution of life expectancy. We can summarize this sample to get point estimates and credible intervals of life expectancy.

Our approach is fully Bayesian in that, in addition to specifying a prior for γ_{ast} , we also specify priors for hyper-parameters, such as σ^2 , that govern the prior for γ_{ast} . Inferences about the hyper-parameters are made together with inferences about γ_{ast} , using the joint posterior distribution. An alternative approach, known as Empirical Bayes, is to construct point estimates for the hyper-parameters and make inferences about γ_{ast} conditional on these point estimates (Rao and Molina, 2015, Chapter 9).

Empirical Bayes approaches can be less computationally intensive than fully Bayesian ones, which means they sometimes scale better to large datasets. They can, however, be difficult to implement with complicated models containing many levels, such as ours. Using probability distributions, rather than point estimates, for hyper-parameters also leads to a more complete representation of uncertainty.

2.3 Results

Figure 2.3 shows results from the model. The light blue band represents 95% credible intervals. If the assumptions of the model are met, then each vertical slice of the band has a 95% probability of containing the true value for γ_{ast} . The pale line in the middle is the median of the posterior distribution, which can be used for point estimates. The black circles are the direct estimates from Figure 2.1.

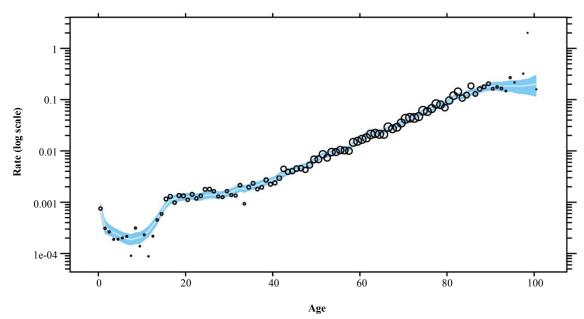


Figure 2.3 Modelled estimates of mortality rates for Māori males in 2012-2014, by single year of age. The light blue band represents 95% credible intervals, and the white line represents posterior medians. The black circles represent direct estimates.

The age pattern obtained from the model is approximately linear over ages 40-80. The model successfully smooths through the random variation in the direct estimates. Around age 18, however, the slope changes abruptly, marking the beginning of the accident hump. The smoothness at ages 40 and over does not come at the expense of an ability to detect local changes in the teenage years. The model also makes no attempt to smooth away the spike in mortality at age 0. This is a result of the inclusion of a covariate for age 0: models that do not have this term do partly smooth away the spike (results not shown).

Estimates around age 10 are, on a log scale, less precise than for most other age groups, reflecting the small cell counts for children. In other words, the model produces uncertainty measures that reflect local availability of data.

Uncertainty also increases steadily beyond age 90, as death counts become smaller and smaller. The posterior median suggests little increase in death rates beyond age 90. The apparent plateauing in mortality rates may be genuine: Māori males who survive to age 90 may systematically differ from ones who do not, so that the flat mortality for people at high ages reflects a kind of selection effect (Vaupel, Manton and Stallard, 1979). However, it is also possible that the plateauing reflects problems with the input data, such as inaccurate recording of ages of very old people.

Figure 2.4 shows life expectancies derived from the model. Unlike Figures 2.1 and 2.3, it shows results for both sexes and both periods. Female life expectancy at birth increased from 75.1 years, with a 95% credible interval of (74.8, 75.5), in 2005-2007 to 76.7 years, with a credible interval of (76.4, 77.0), in

2012-2014. The corresponding estimates for males are 70.8 (70.5, 71.1) and 72.5 (72.2, 72.9). It is still rare in demography for life expectancies to be accompanied by uncertainty measures. Using Bayesian methods, however, uncertainty measures can be calculated routinely.

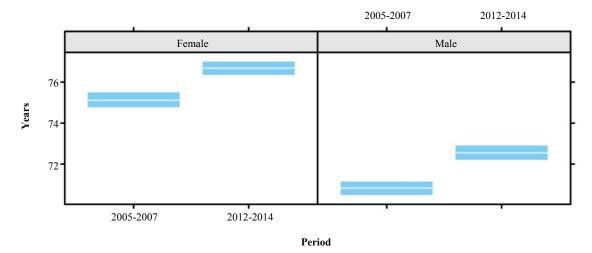


Figure 2.4 Modelled estimates of life expectancy at birth, for Māori, 2005-2007 and 2012-2014. The light blue bands represent 95% credible intervals, and the white lines represent posterior medians.

3 Interpolating and forecasting obesity prevalence

3.1 The estimation problem

In New Zealand, as in most countries, obesity rates are rising. Public health researchers and policy makers monitor and forecast obesity prevalence, to assess the success, or otherwise, of obesity-reduction measures, and to gauge future demand for services.

The main source of data on obesity prevalence in New Zealand is the New Zealand Health Survey, a nationally-representative survey of around 19,000 people (Ministry of Health, 2013). Like most household surveys, it has a complex design, with stratification and clustering. Obesity is measured using body mass index (BMI). A person is defined as being obese if he or she has a BMI of 30 or higher.

Surveys were carried out in 1997, 2003, 2007, 2012, and 2013. We use data for all these years. Our objective is to obtain prevalence estimates for the period 1997-2013, including non-survey years, and then forecast for the period 2014-2023. Our estimates and forecasts are disaggregated into age groups 15-24, 25-34, 35-44, 45-54, 55-64, 65-74, and 75+.

3.2 The model

Our main input data are published estimates for the proportion of New Zealanders aged *a* at time *t* who are obese, which we denote p_{at} , and the published standard errors for the p_{at} , denoted s_{at} . The p_{at} are graphed in Figure 3.1.

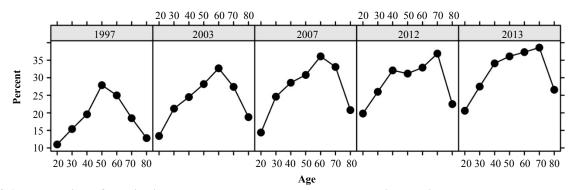


Figure 3.1 Proportion of obesity in New Zealand, by age and year, as estimated in the New Zealand Health Survey.

When individual-level data are available, the standard Bayesian approach towards accounting for complex survey design is to include as many features of the design as possible in the estimation model (Gelman et al., 2014, Chapter 8). Chen, Wakefield and Lumely (2014) show, however, how the full individual-level approach can be approximated by an aggregate-level approach that starts from design based estimates such as p_{at} and s_{at} . Chen et al. (2014) assume that the design-based estimates are constructed so as to reflect all the important features of the survey design, and show how these estimates can be converted into a form suitable for inclusion in an aggregate-level model.

Applying the approach of Chen et al. (2014), we approximate the individual-level approach using a Binomial likelihood. We obtain counts of individuals with obesity y_{at} and total counts of individuals n_{at} by finding y_{at} and n_{at} such that $\frac{y_{at}}{n_{at}} \approx p_{at}$ and $\frac{y_{at}}{n_{at}} \left(1 - \frac{y_{at}}{n_{at}}\right) \approx s_{at}^2$. The likelihood is

$$y_{at} \sim \text{Binomial}(n_{at}, \gamma_{at}).$$
 (3.1)

Here γ_{at} is the super-population probability of obesity: the probability that a person aged *a* at time *t* is obese. Our objective is to estimate γ_{at} for past years, including years without survey data, and to forecast γ_{at} for future years.

Our prior model for γ_{at} is

$$logit(\gamma_{at}) = \beta^0 + \beta_a^{age} + \beta_t^{time}, \qquad (3.2)$$

which includes age and time effects, but not an age-time interaction. We experimented with an age-time interaction, but found that its size was small enough to omit (results not shown).

As with the mortality model of Section 2, we use a local trend model for the age effect, though in the obesity case we do not have an infant covariate. The rationale for using a local trend model is, once again, to capture the correlations between neighbouring age groups. We also use the same prior for the intercept as we do in Section 2, a normal distribution with mean 0 and standard deviation 10.

We use a local trend model for time,

$$\beta_t^{\text{time}} = \alpha_t^{\text{time}} + u_t^{\text{time}} \tag{3.3}$$

$$\alpha_t^{\text{time}} = \alpha_{t-1}^{\text{time}} + \delta_{t-1}^{\text{time}} + v_t^{\text{time}}$$
(3.4)

$$\delta_t^{\text{time}} = \delta_{t-1}^{\text{time}} + w_t^{\text{time}}, \qquad (3.5)$$

but with two different sets of assumptions about innovation terms v_t^{time} and w_t^{time} .

In our first version, we assume that v_t^{time} and w_t^{time} are always very close to 0, which we implement by using extremely tight priors on the standard deviations for these terms. The standard deviations for both terms have half-t priors with scales of 0.001. This version of the local trend model essentially fits a straight line through the data. Aside from assuming no change, this is perhaps the most common approach to forecasting future rates in epidemiology and demography. We refer to this model as the "straight line" model.

Our second version is a generalization of the first. Rather than assuming that v_t^{time} and w_t^{time} are always close to 0, we allow them to take values that imply year-on-year changes in obesity rates of a few percentage points. We do this by setting the scale of the prior for the standard deviation of v_t^{time} to 0.05 and setting the scale of the prior for standard deviation of w_t^{time} to 0.025. We use a larger scale for v_t^{time} than for w_t^{time} on the basis that levels change more rapidly than systematic trends. We refer to the model based on this version of the time effect as the "flexible" model.

We carry out the estimation using our package *demest*, with the same settings for burnin, production, chains, and thinning as for the mortality application.

3.3 Results

Figure 3.2 shows results based on the "straight line" model. Estimates for survey years are shown in red, and estimates and forecasts for the remaining years are shown in blue. As is conventional with forecasting, we use 80% credible intervals, rather than 95%.

Estimates for years with survey data are more precise than those for years without survey data, as we would expect. Estimates for years between surveys are more precise than those for forecasts. The differences in precision between estimates and forecasts are, nevertheless, small. Strong assumptions about linearity lead to precise forecasts.

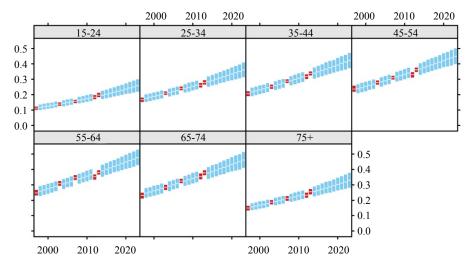


Figure 3.2 Estimates and forecasts of obesity prevalence in New Zealand–"straight line" model. The bands represent 80% credible intervals, and the pale lines represent posterior medians. The red bands are for years with survey data and the light blue bands are for years without data.

Figure 3.3 shows results based on the "flexible" model. Point estimates and forecasts from the flexible model are indistinguishable from those of the straight line model. The level of uncertainty, however, is clearly different. Compared with the straight line model, there is a modest increase in uncertainty for years between surveys and a large increase in uncertainty for the forecast period, particularly in later years.

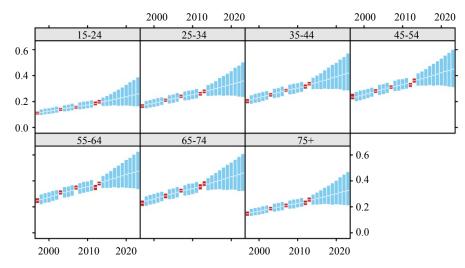


Figure 3.3 Estimates and forecasts of obesity prevalence in New Zealand–"flexible" model. The bands represent 80% credible intervals, and the pale lines represent posterior medians. The red bands are for years with survey data and the light blue bands are for years without data.

The flexible model, arguably, gives a better representation of knowledge about obesity trends in New Zealand than the linear model. The linear assumption is conventional, but does not have any strong theoretical basis. Over-reliance on the linear assumption can produce over-confidence. The flexible model illustrates the implications of weaker assumptions.

4 Discussion

Despite the increasing popularity of Bayesian methods in the research community, national statistical agencies and policy analysts have been wary of these methods. National statistical agencies are particularly concerned about two aspects of Bayesian methods: their use of prior distributions, and their complexity.

The use of prior distributions to represent external information is indeed a distinctive feature of Bayesian analyses. Little (2012) has argued that national statistical agencies should use "noninformative" priors, which avoid the impression of subjectivity, and which form a bridge to classical methods, in that they often lead to similar results. Among Bayesian statisticians, however, weakly informative priors have been gradually displacing noninformative priors as the default for most analyses. Compared with noninformative priors, weakly informative priors can stabilize estimates, and speed up calculations. But because they rule out only the most implausible values, they are generally no more controversial, and require little more work or justification, than noninformative priors.

However, in cases where the data to hand do not permit sufficiently strong answers to the questions of interest, there may be advantages to using priors that are strongly, rather than weakly, informative. In our obesity example, for instance, it may be possible to improve on both of our forecasting models by specifying priors for the standard deviation parameters that accurately reflect likely year-on-year variation in obesity rates.

If statistical agencies were to use strongly informative priors, they would need to spell these priors out clearly, justify their choices, and test sensitivity to alternative choices. But, in most cases, this would be an improvement on current practice. Current practice with analyses such as population forecasts is often to apply informal adjustments, or to retrospectively adjust assumptions, until a plausible result is obtained. Bayesian methods provide analysts with a more transparent and systematic way of bringing in external information and expert judgement.

Objections about Bayesian models being complicated are partly true. Many Bayesian models *are* complicated, in that, like the models presented in this paper, they use many layers and many parameters. At the same time, however, the individual components of these models are often simple and intuitive. To make sense of our model for mortality rates, for instance, we can start with the likelihood, move on to the prior model, and then consider the priors for main effects and interactions one by one. With this divide-and-conquer approach, even complicated models are accessible. Moreover, the main assumptions behind the models can often be described in nontechnical language, even if the mathematical techniques cannot.

Similarly, the traditional objection that Bayesian modelling require advanced computing skills is gradually losing force. Packages such as ours allow analysts to fit specific classes of demographic estimation models relatively easily. General-purpose Bayesian programming languages such as Stan (Carpenter, Gelman, Hoffman, Lee, Goodrich, Betancourt, Brubaker, Guo, Li and Riddell, 2016) offer greater flexibility in exchange for slightly more programming effort. These tools allow practitioners to easily fit complicated Bayesian models.

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Appendix A

Identification of our model

In the prior model, each main effect or interaction includes all possible categories of the classifying dimensions. For instance, the sex effect includes separate female and male effects, and the age-sex interaction includes effects for every possible combination of age and sex. Because all of our priors are proper (i.e., are genuine probability distributions that integrate to 1), the posterior distribution is proper. All parameters are therefore identified in the broadest sense, and a Bayesian analysis can be carried out.

The main effects and interactions are, however, only weakly identified. For instance, adding a value λ to the female and male effects $\beta_{\text{female}}^{\text{sex}}$ and $\beta_{\text{Male}}^{\text{sex}}$, and subtracting λ from the intercept β^0 , will produce

exactly the same expected value for the γ_{ast} as the original parameter settings. The data do not allow us to distinguish between the two possibilities. Identification is achieved entirely through the differences in prior densities for the original and shifted parameters.

The γ_{ast} , in contrast, cannot be arbitrarily shifted without affecting the likelihood Poisson $(y_{ast} | \gamma_{ast} n_{ast})$. In others words, the γ_{ast} are well identified from the data. Shifting the values of the main effects and interactions does not affect inferences about standard deviation terms, as inferences about standard deviations depend on variation across effects, rather than absolute levels. The standard deviation terms are therefore also well identified.

In this paper we only report the γ_{ast} . In some applications, however, the main effects and interactions are also of interest. In such cases, one approach is to systematically shift the parameter estimates to achieve identification (Gelman, 2005).

Appendix B

R code

We have developed a set of R packages for implementing Bayesian small area demographic estimation and forecasting. The packages are available at github.com/statisticsnz/R. Package *dembase* contains data structures for demographic data and functions for manipulating these data structures. The basic data structure is a "demographic array", which, in addition to the counts or rates themselves, also holds metadata such as age groups or time periods, and units of measurement. Bayesian estimation and forecasting is carried out by functions in package *demest*. The estimation functions use metadata from the demographic arrays to assign sensible default values. As a result, complex models can be specified and run relatively simply. For instance, the key parts of the code for our model in the mortality example are set out in Figure B.1. Package *demlife* contains tools for creating life tables and extracting life table functions.

Figure B.1 R code to specify and run the mortality model, using package demest.

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Small area estimation of survey weighted counts under aggregated level spatial model

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Abstract

The empirical predictor under an area level version of the generalized linear mixed model (GLMM) is extensively used in small area estimation (SAE) for counts. However, this approach does not use the sampling weights or clustering information that are essential for valid inference given the informative samples produced by modern complex survey designs. This paper describes an SAE method that incorporates this sampling information when estimating small area proportions or counts under an area level version of the GLMM. The approach is further extended under a spatial dependent version of the GLMM (SGLMM). The mean squared error (MSE) estimation for this method is also discussed. This SAE method is then applied to estimate the extent of household poverty in different districts of the rural part of the state of Uttar Pradesh in India by linking data from the 2011-12 Household Consumer Expenditure Survey collected by the National Sample Survey Office (NSSO) of India, and the 2011 Indian Population Census. Results from this application indicate a substantial gain in precision for the new methods compared to the direct survey estimates.

Key Words: Complex surveys; Direct survey weighted estimator; Poverty estimate; Spatial Model; Mapping.

1 Introduction

Sample surveys are generally planned to produce estimates for population characteristics of interest mainly at higher geographic (e.g., national and state) levels. The sample size is fixed in such a way that the direct estimators for larger domains provide reliable estimates, where by direct estimators we mean estimators that use only sample-weighted data from the domain of interest. In many practical situations, however, the aim is to estimate parameters for domains that contain only a small number of sample observations. The term "small areas" is used to describe domains whose sample sizes are not large enough to allow sufficiently precise direct estimation. When direct estimation is not possible, one has to rely on alternative, model-based methods for producing small area estimates. Such methods depend on model specification as well as on the availability of population level auxiliary information related to the variable of interest, and are commonly referred to as indirect methods (Rao, 2003). The underlying theory is referred to as the small area estimation (SAE), and SAE techniques aim at producing reliable estimates based on such small sample sizes by using the model "linking" the small areas to "borrow strength" from the sample data from other small areas, see for example, Pfeffermann (2002) and Rao and Molina (2015). In this context, we differentiate between SAE methods based on unit-level models and those based on area-level models. In the former case these models are for the individual survey measurements and include area effects, while in the latter case these models are used to smooth out the variability in the unstable area-level direct survey estimates. Area-level modelling is typically used when unit-level data are unavailable, or, as is often the case, where model covariates (e.g., census variables) are only available at area level. The Fay-Herriot model (Fay and Herriot, 1979), is a widely used area level model that assumes area-specific survey estimates

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are available, and that these follow an area level linear mixed model with independent area random effects. This model can also accommodate survey weights in SAE by using the survey weighted direct estimates when fitting the linear mixed model. When the variable of interest is not continuous (for example, binary and count data), a generalized linear mixed model (GLMM) is often used. If the variable of interest is binary and the target of inference is a small area proportion, then the GLMM with logistic link function (also referred as the logistic linear mixed model) is commonly used. An empirical plug-in predictor (EP) is commonly used for the estimation of small area proportions under a GLMM, see for example, Chandra, Chambers and Salvati (2012), Salvati, Chandra and Chambers (2012), Rao and Molina (2015) and references therein, although it is not the most efficient predictor under this model. An alternative is the Empirical Best Predictor (EBP, Jiang, 2003). This predictor does not have a closed form in general and so must be computed via numerical approximation, which is typically not straightforward. As a consequence national statistical agencies tend to favour computation of an approximation like the EP. It is also our understanding that EP-type predictors are used in Molina, Saei and José Lombardia (2007) and Lopez-Vizcaino, Lombardìa and Morales (2013). When only area level data are available, an area level version of a GLMM can be considered for SAE, see for example, Saei and Chambers (2003), Johnson, Chandra, Brown and Padmadas (2010), Chandra, Salvati and Sud (2011), Chandra, Salvati and Chambers (2017) and references therein. Unlike the Fay-Herriot model, this approach implicitly assumes simple random sampling with replacement within each area and ignores the survey weights. Unfortunately, this has the potential to seriously bias the estimates if the small area samples are seriously unbalanced with respect to key population charcteristics. Consequently use of the survey weights appears to be inevitable for if one wishes to generate representative small area estimates.

Modelling the survey weighted estimates of proportions by a continuous distribution, as in the Fay-Herriot model, can be problematic since the sampling distribution of these proportions is inherently discrete due to small sample sizes and binary nature of the underlying observations. For example, such a linear model based approach can lead to negative predictions or, more likely, prediction intervals that include negative values, both absurd results. Clearly GLMMs are more suitable for modelling inherently discrete data arising from survey weighted direct estimates, see Liu, Lahiri and Kalton (2014), Ghosh, Natarajan, Stroud and Carlin (1998) and Rao (2003, Sections 5.6 and 10.11). Malec, Sedransk, Moriarity and LeClere (1997) describe a hierarchical Bayes model for binary survey data. The authors examined the use of sampling weights as a covariate in the model and did not find any improvement for their example of county-level data from the National Health Interview Survey. Unlike estimation of survey weighted linear parameters like small area means and totals, there has been comparatively little research on estimation of survey weighted small area proportions or counts under area level small area models. Recently, some authors including Mercer, Wakefield, Chen and Lumley (2014), Liu et al. (2014) and Franco and Bell (2013) have described how survey weights can be used in a Bayesian hierarchical model framework for estimating small area proportions by using effective sample sizes for inference about the underlying binomial distributions in order to preserve sampling variances estimated via a generalized variance function. These approaches are defined within a Bayesian framework. In this article, we adopt a frequentist approach and describe an SAE method that uses the sampling weights for binary data when estimating small area proportions under an area level version of a GLMM. Following Korn and Graubard (1998), we model the survey weighted estimates as binomial proportions, with an "effective sample size" chosen to match the binomial variance to the sampling variance of the estimates. Using the effective sample size rather than the actual sample size allows for the varying information in each area under complex sampling. Furthermore, the area-specific random effects in the GLMM are typically assumed to be independent, that is, different small areas are considered to be independent of each other. However, in practice most small area boundaries are arbitrary and there appears to be no good reason why units that are just one side of such a boundary should not generally be correlated with units just on the other side (Pratesi and Salvati, 2008). One approach to incorporating such spatial information in SAE modelling is to extend the random effects model to allow for spatially correlated area effects using, for example, a Simultaneous Autoregressive (SAR) model. See Anselin (1992) and Cressie (1993). These models embed spatial behaviour in the model random effects, which in the context of SAE typically means area effects. Applications of SAR models in small area estimation have been considered by Singh, Shukla and Kundu (2005), Pratesi and Salvati (2008), Pratesi and Salvati (2009), Molina, Salvati, and Pratesi (2009), Marhuenda, Molina and Morales (2013) and Porter, Wikle and Holan (2015). SAR models are commonly used for modelling spatial dependence under the frequentist approach to SAE. Other the hand, Bayesian approaches to SAE favour Conditionally Auto-Regressive (CAR) models for modelling spatial dependence, see Besag, York and Mollié (1991), Leroux, Lei and Breslow (1999) and Mercer et al. (2014). This paper takes a frequentist approach to SAE and so we consider an extension of an area level GLMM to account for spatial dependence between the small areas based on a SAR specification, and develop small area estimates under this model that use the survey design information.

The structure of the paper is as follows. Section 2 describes the data from the 2011-12 Household Consumer Expenditure Survey of the National Sample Survey Office (NSSO) of India and the 2011 Indian Population Census that will be used to estimate the district level incidence of household poverty in the rural part of the Indian State of Uttar Pradesh. In Section 3 we set out the theoretical background to the area level version of the GLMM, and then discuss the use of effective sample size for incorporating the information in the survey weights when estimating small area proportions under this model. The spatial extension of the area level GLMM, as well as an empirical predictor based on this model, are also introduced in this section. The results from the poverty incidence application along with various diagnostic measures are reported in Section 4. In this section we also provide a poverty map that serves to demonstrate district-level inequalities in the distribution of poor rural households in Uttar Pradesh. Finally, Section 5 summarizes the paper and provides concluding remarks.

2 Data description

This section introduces the basic sources of the data, i.e., the 2011-12 Household Consumer Expenditure Survey (HCES) of the NSSO for rural areas of the State of Uttar Pradesh in India and the 2011 Population Census, used in the small area application reported in this paper. Data obtained from these sources are then

used to estimate the proportion of poor households at district level in Uttar Pradesh. The State of Uttar Pradesh is the most populous State in the country and accounts for about 16.16 per cent of India's population. It covers 243,290 square km, equal to 6.88% of the total area of the country. Poverty estimates in India are produced for all the States separately for both rural and urban sectors. Our analysis is restricted to the rural areas of Uttar Pradesh because about 78% of the population of this State live in rural areas according to 2011 Population Census. The NSSO conducts nationwide HCE surveys at regular intervals as part of its "rounds", with the duration of each round normally being a year. These surveys are aimed at generating estimates of average household monthly per capita consumer expenditure (MPCE), the distribution of households and persons over the MPCE range, and the break-up of average MPCE by commodity group, separately for the rural and urban sectors of the country, for States and Union Territories, and for different socio-economic groups. These indicators are amongst the most important measures of the living conditions of the relevant domains of the population. The surveys are conducted through interviews of a representative sample of households selected randomly through a suitable sampling design and covering almost the entire geographical area of the country. In particular, the sampling design used in the NSSO survey is stratified multi-stage random sampling with districts as strata, villages as first stage units and households as second stage units. Although, these surveys provide reliable and representative state and national level estimates, they cannot be used directly to produce reliable estimates at the district level due to small sample sizes. In particular, although district is a very important domain of the planning process in India, there are no surveys aimed at producing estimates at this level. The lack of robust and reliable outcome measures at the district level puts constraints on the design of targeted interventions and policy development. More importantly, state and national estimates do not adequately capture the extent of geographical inequalities, which restricts the scope for evaluating progress locally within and between districts. Balanced against all of this however is the fact that conducting a district level survey would be very costly as well as time-consuming. In the 2011-12 HCES, a total of 5,916 households from the 71 districts of Uttar Pradesh were surveyed. The district sample sizes ranged from 32 to 128 with average of 83. It is evident that these district level sample sizes are relatively small, with an average sampling fraction of 0.0002. As a consequence, it is difficult to generate reliable district level direct survey estimates with associated standard errors from this survey. This small sample size problem can be resolved by using SAE methodology provided auxiliary information is available to strengthen the limited sample data from the districts (Rao and Molina, 2015; Tzavidis, Salvati, Pratesi and Chambers, 2008).

We note here that the target variable Y at the unit (household) level in the published survey data file is binary, corresponding to whether a household is poor or not. In our application however we focus on estimation where the available data are the corresponding counts of the number of poor households in sample in each district. In this context a household having MPCE below the state poverty line is defined as being poor. The poverty line used in this study (Rs. 768) is the same as that set by the Planning Commission, Govt. of India, for 2011-12. The parameter of interest is then the proportion of poor rural households within each district. The auxiliary variables (covariates) used in our analysis are taken from the Indian Population Census of 2011. These auxiliary variables are only available as counts at district level, and so SAE methods based on area level small area models, as described in next section, must be employed to derive the small area estimates. There are approximately 50 such covariates that are available for use in SAE analysis. We therefore carried out a preliminary data analysis in order to define appropriate covariates for SAE modelling, using Principal Component Analysis (PCA) to derive composite scores for selected groups of variables. In particular, we carried out PCA separately on three groups of variables, all measured at district level and identified as S1, S2 and S3 below. The first group (S1) consisted of literacy rates by gender and proportions of worker population by gender. The first principal component (S11) for this group explained 51% of the variability in the S1 group, while adding the second principal component (S12) increased explained variability to 85%. The second group (S2) consisted of the proportions of main worker by gender, proportions of main cultivator by gender and proportions of main agricultural labourer by gender. The first principal component (S21) for this second group explained 49% of the variability in the S2 group, while adding the second component (S32) increased explained variability to 67%. Finally, the third group (S3) consisted of proportions of marginal cultivator by gender and proportions of marginal agriculture labourers by gender. The first principal component (S31) for this third group explained 61% of the variability in the S3 group, while adding the second component (S22) increased explained variability to 78%. Using the methods detailed in the following sections, we fitted a generalised linear model using direct survey estimates of proportions of poor rural households as the response variable and the six principal component scores S11, S12, S21, S22, S31 and S32 as potential covariates. The final selected model included the three covariates S11, S21 and S31, with residual deviance and AIC values of 327.18 and 636.89, respectively. This final model was then used to produce district wise estimates of rural poverty incidence, i.e., estimates of the head count ratio (HCR) at this level.

A major problem with application of SAE in many developing and underdeveloped countries is the fact that administrative or civic registration data that are suitable for use as covariates in the SAE model are unavailable at small area level. Typically, what auxiliary variables are available (e.g., census tabulations) do not have good association with variable of interest. This results in very limited information for producing the small area estimates. In such cases, it can be beneficial to supplement the available data by using geospatial information about the small areas. In this study, we use the geographical locations (centroids) of the different districts in the Indian State of Uttar Pradesh to extend our SAE model to one that allows rural poverty counts from neighbouring districts to be correlated.

3 Small area estimation under the area level GLMM

3.1 Spatially uncorrelated random area effects

We assume that a probability sampling method is used to draw a sample s of size n from a finite population U of size N, which consists of D non-overlapping domains U_i (i = 1, ..., D). Following standard practice, we refer to these domains as small areas or just areas. Furthermore, we assume that there is a known number N_i of population units in small area i, with n_i of these sampled. The total number of

units in the population is $N = \sum_{i=1}^{D} N_i$, with corresponding total sample size $n = \sum_{i=1}^{D} n_i$. We use s to denote the collection of units in sample, with s_i the subset drawn from small area i (i.e., $|s_i| = n_i$), and use expressions like $j \in i$ and $j \in s$ to refer to the units making up small area i and sample s, respectively. Similarly, r_i denotes the set of units in small area *i* that are not in sample, with $|r_i| = N_i - n_i$ and $U_i = s_i \cup r_i$. Let y_{ij} denotes the value of the variable of interest for unit j $(j = 1, ..., N_i)$ in area i. The variable of interest, with values y_{ij} , is binary (e.g., $y_{ij} = 1$ if household j in area i is poor household and 0 otherwise), and the aim is to estimate the small area population count, $y_i = \sum_{j \in U_i} y_{ij}$, or equivalently the small area proportion, $P_i = N_i^{-1} \sum_{j \in U_i} y_{ij}$, in area i (i = 1, ..., D). The standard direct survey estimator (hereafter denoted by DIR) for P_i is $p_{iw} = \sum_{j \in s_i} \tilde{w}_{ij} y_{ij}$, where $\tilde{w}_{ij} = w_{ij} / \sum_{j \in s_i} w_{ij}$ is the normalized survey weight for unit j in area i with $\sum_{j \in s_i} \tilde{w}_{ij} = 1$ and w_{ij} is the survey weight for unit j in area i. The estimated design-based variance of DIR is approximated by $v(p_{iw}) \approx$ $\sum_{i \in s_i} \tilde{w}_{ij} (\tilde{w}_{ij} - 1) (y_{ij} - p_{iw})^2$. This formula for the variance estimator of DIR is obtained from Särndal, Swensson and Wretman (1992; see pages 43, 185 and 391), with the simplifications $w_{ij} = a_{ij}^{-1}$, $a_{ij,ij} = a_{ij}$ and $a_{ij,ik} = a_{ij}a_{ik}$, $j \neq k$, where a_{ij} is the first order inclusion probability of unit j in area i and $a_{ij,ik}$ is the second order inclusion probability of units j and k in area i. Under simple random sampling (SRS), $w_{ii} = N_i n_i^{-1}$ and DIR is then $p_i = n_i^{-1} y_{si}$, with estimated variance $v(p_i) \approx n_i^{-1} p_i (1 - p_i)$, where $y_{si} = \sum_{j \in s_i} y_{ij}$ denotes the sample count in area *i*.

If the sampling design is informative, the SRS-based version of DIR defined in the previous paragraph may be biased. For the 2011-12 HCES we therefore computed district specific design effects, defined by the ratios of the variance of the weighted estimates (i.e., for a given sampling design) to the variance of the unweighted estimates (i.e., assuming SRS). These design effects were more than one in all but three districts, with values greater than one that varied from 1.17 to 8.44 with an average of 2.71. This is strong evidence that the sampling design used in the 2011-12 HCES is informative. Furthermore, DIR is based on area-specific sample data and can therefore be very imprecise when the area specific sample size is small or may even be impossible to compute if this sample size is zero. However, model-based SAE procedures that "borrow strength" via a common statistical model for all the small areas can be used to address this problem, see Rao and Molina (2015).

Suppose now that the available data consist of the sample aggregates y_{si} (i.e., the sample counts of poor households), together with the values of area specific contextual covariates. That is, for area *i* we observe the count y_{si} together with a *k*-vector of area-specific covariates \mathbf{x}_i derived from secondary data sources (e.g., the census or administrative registers). If we ignore the sampling design, the sample count y_{si} in area *i* can be assumed to follow a Binomial distribution with parameters n_i and π_i , i.e., $y_{si} \sim \text{Bin}(n_i, \pi_i)$, where π_i is the probability of occurrence of an event for a population unit in area *i* or the probability of prevalence in area *i*. Following Saei and Chambers (2003), Johnson et al. (2010) and Chandra et al. (2011), the model linking the probability π_i with the covariates \mathbf{x}_i is the logistic linear mixed model of form

$$\operatorname{logit}(\pi_i) = \ln\left\{\pi_i \left(1 - \pi_i\right)^{-1}\right\} = \eta_i = \mathbf{x}_i^T \mathbf{\beta} + u_i, \qquad (3.1)$$

with $\pi_i = \exp(\mathbf{x}_i^T \mathbf{\beta} + u_i) \{1 + \exp(\mathbf{x}_i^T \mathbf{\beta} + u_i)\}^{-1} = \exp((\mathbf{x}_i^T \mathbf{\beta} + u_i))$. Here $\mathbf{\beta}$ is the *k*-vector of regression coefficients, often referred to as the vector of fixed effects, and u_i is the area-specific random effect, with $u_i \sim N(0, \sigma_u^2)$. The model (3.1) is a special case of the generalized linear mixed model (GLMM) with a logit link function (Breslow and Clayton, 1993). We can observe that the parameters $\mathbf{\beta}$ and σ_u^2 are the same for every area; i.e., they can be estimated using the data from all small areas. This is usually accomplished by "stacking" the *D* area level models given by (3.1) to produce a population level model of the form

$$\eta = \mathbf{X}\boldsymbol{\beta} + \mathbf{Z}\mathbf{u} \tag{3.2}$$

with $\boldsymbol{\pi} = (\pi_1, ..., \pi_D)^T = \operatorname{expit}(\boldsymbol{\eta})$, where $\boldsymbol{\eta} = (\eta_1, ..., \eta_D)^T$, $\mathbf{X} = (\mathbf{x}_1^T, ..., \mathbf{x}_D^T)^T$ is the $D \times k$ matrix of covariates, $\mathbf{Z} = (\mathbf{z}_1, ..., \mathbf{z}_D)^T$ is a matrix of known covariates of dimension $D \times D$ characterising differences among the small areas, \mathbf{z}_i is the *D*-vector $(0, ..., 1, ..., 0)^T$ with the 1 in the *i*th position and $\mathbf{u} =\geq (u_1, ..., u_D)^T$ is the *D*-vector of random area effects with $\mathbf{u} \sim N(\mathbf{0}, \boldsymbol{\Omega})$ where $\boldsymbol{\Omega} = \boldsymbol{\Omega}(\sigma_u^2) = \sigma_u^2 \mathbf{I}_D$ and \mathbf{I}_D is a $D \times D$ diagonal matrix.

The Penalized Quasi-Likelihood (PQL) approach is a widely used estimation procedure when fitting a GLMM. This approach constructs a linear approximation to the non-normal response variable and then assumes that this linearized dependent variable is approximately normal. The PQL method is widely used in small area estimation because it is much easier to implement in practice than estimation under the EBP approach. In particular, we employ a hybrid approach, with PQL used to estimate the parameters β and **u** in the GLMM (3.2), and restricted maximum likelihood used to estimate the variance parameter $\Omega(\sigma_u^2)$. See Breslow and Clayton (1993), Saei and Chambers (2003) and Manteiga, Lombardia, Molina, Morales and Santamarìa (2007). It is known that in some situations PQL can lead to inconsistent estimators, however recent empirical applications of PQL (Manteiga et al., 2007) note that the method works well in practice.

Under the model (3.2), the expected values of y_{si} and y_{ri} given u_i are given by $\mu_{si} = E(y_{si} | u_i) = n_i \exp((\mathbf{x}_i^T \boldsymbol{\beta} + \mathbf{z}_i^T \mathbf{u}))$ and $\mu_{ri} = E(y_{ri} | u_i) = (N_i - n_i) \exp((\mathbf{x}_i^T \boldsymbol{\beta} + \mathbf{z}_i^T \mathbf{u}))$, respectively. Under (3.2), a plug-in empirical predictor (EP) of the population count y_i in area *i* is

$$\hat{y}_{i}^{\text{EP}} = y_{si} + \hat{\mu}_{ri} = y_{si} + (N_{i} - n_{i}) \hat{\pi}_{i}^{\text{EP}}, \qquad (3.3)$$

where $\hat{\pi}_i^{\text{EP}} = \exp((\mathbf{x}_i^T \hat{\mathbf{\beta}} + \mathbf{z}_i^T \hat{\mathbf{u}}))$. An estimate of the corresponding proportion in area *i* is obtained as $\hat{P}_i^{\text{EP}} = N_i^{-1} \hat{y}_i^{\text{EP}}$. For non-sampled area *i* with associated vector of covariates $\mathbf{x}_{i,\text{out}}$, the synthetic estimator of y_i under (3.2) is $\hat{y}_i^{\text{SYN}} = N_i \hat{\pi}_i^{\text{SYN}}$, with $\hat{\pi}_i^{\text{SYN}} = \exp((\mathbf{x}_{i,\text{out}}^T \hat{\mathbf{\beta}}))$.

The model (3.1), being based on unweighted sample counts, assumes that sampling within areas is noninformative given the values of the contextual variables and the random area effects. As a result, the predictor (3.3) ignores the complex survey design. If the sampling design is informative and survey weighted counts are available, there are two main difficulties. First, the values for the weighted sample counts will not necessarily be the integers $0, 1, ..., n_i$; rather they will take a value from a finite set of unequally-spaced numbers (not necessarily integers) determined by the survey weights of the sample cases in area *i*. Second, the estimated sampling variance of the weighted sample counts, y_{si} implied by the Binomial distribution, i.e., $v(y_{siw}) \approx n_i p_{iw} (1 - p_{iw})$, will be incorrect. Korn and Graubard (1998) suggest that one instead model the survey weighted probability estimate for an area as a binomial proportion, with an "effective sample size" that equates the resulting binomial variance to the actual sampling variance of the survey weighted direct estimate for the area. The use of "effective sample size" has been discussed by a number of authors including Mercer et al. (2014) and Liu et al. (2014) as a way of incorporating the survey weights. Mercer et al. (2014) observe that the pseudo likelihood approach and effective sample size approach lead to identical estimates of small area proportions. Using the effective sample size rather the actual sample size allows for the survey weights under complex sampling. Furthermore, the precision of an estimate from a complex sample can be higher than for a simple random sample, because of the better use of population data through a representative sample drawn using a suitable sampling design. Here we use a subscript of (e) in all the quantities associated with the "effective sample size". We address the above two issues by defining an "effective sample size" $n_{i(e)}$, and an "effective sample count" $y_{is(e)}$, such that $y_{is(e)} = n_{i(e)}p_{iw}$. This leads to $p_{iw} = \frac{y_{is(e)}}{n_{i(e)}}$ with its corresponding estimator of variance estimate $v(p_{iw})$. The model (3.1) is then applied assuming the effective sample count $y_{is(e)}$ in small area *i* follows a Binomial distribution, i.e., $y_{is(e)} \sim Bin(n_{i(e)}, \pi_i)$. The "effective sample size" $n_{i(e)}$ is given by $n_{i(e)} = \frac{\hat{P}_i(1-\hat{P}_i)}{v^*(p_{iw})}$, where \hat{P}_i is a preliminary model-based prediction of the population proportion P_i under a generalised linear model, and the estimate of variance $v^*(p_{iw})$ depends on \hat{P}_i through a fitted generalized variance function (GVF), see Liu et al. (2014). Here, $y_{is(e)} = 0$ if $p_{iw} = 0$, but this does not cause problems since $\hat{P}_i > 0$ implies $n_{i(e)} > 0$. Note that we use a generalised variance function (GVF) to generate estimates of the sampling variance even for areas that have an observed count of zero. Consequently we do not exclude any area from model fitting. The empirical predictor of y_i is finally obtained by replacing (n_i, y_{si}) by $(n_{i(e)}, y_{is(e)})$ in (3.2), thus ensuring that sampling weights are used in the small area estimation process. In particular, the plug-in empirical predictor (EP) of y_i is then

$$\hat{y}_{i(e)}^{\text{EP}} = y_{is(e)} + \left(N_i - n_{i(e)}\right)\hat{\pi}_{i(e)}^{\text{EP}}, \qquad (3.4)$$

with $\hat{\pi}_{i(e)}^{\text{EP}} = \exp i \left(\mathbf{x}_{i}^{T} \hat{\mathbf{\beta}}_{(e)} + \mathbf{z}_{i}^{T} \hat{\mathbf{u}}_{(e)} \right)$. The estimate of the proportion in area *i* is $\hat{P}_{i(e)}^{\text{EP}} = N_{i}^{-1} \hat{y}_{i(e)}^{\text{EP}}$. Here $\hat{\mathbf{\beta}}_{(e)}$ and $\hat{u}_{i(e)}$ are the estimates of the fixed effects parameter and the predictor of the random effects parameter respectively under model (3.2), based on an "effective sample size" $n_{i(e)}$, and an "effective sample count" $y_{is(e)}$. Similarly, the synthetic estimator of y_i is $\hat{y}_{i(e)}^{\text{SYN}} = N_i \hat{\pi}_{i(e)}^{\text{SYN}}$, with $\hat{\pi}_{i(e)}^{\text{SYN}} = \exp i \left(\mathbf{x}_{i,\text{out}}^{T} \hat{\mathbf{\beta}}_{(e)} \right)$.

3.2 Computation of effective sample size

Building on the ideas set out in Liu et al. (2014) and Franco and Bell (2013), we describe a procedure for calculating the district-wise values of effective sample size as well as the corresponding effective sample count of the number of poor households. We first obtain an approximate model-based prediction of P_i , say \hat{P}_i , from a logistic linear model fitted to district-specific direct (i.e., weighted) estimates p_{iw} and a set of district level auxiliary variables. This model is fitted using the *glm* function in R, specifying the family as "binomial" and with the district specific sample sizes as the weights. By definition, these model-based estimates, $\hat{P}_i = \exp(\mathbf{x}_i^T \hat{\boldsymbol{\lambda}})$ will lie inside the interval (0, 1). Here \mathbf{x}_i denotes the vector of covariates used in *glm* function, as in model (3.1). We then use the direct variance estimates $s_i^2 = v(p_{iw})$ from the NSSO survey as the dependent variables in a GVF model. Using the data from districts that do not have zero poverty counts, we develop a smoothed estimate of the sampling variance through the following model:

$$E(s_i^2) = \text{GVF}_i = \alpha_0 \left[P_i \left(1 - P_i \right) \right]^{\alpha_1} \left(R_i \right)^{\alpha_2}.$$

Here $R_i = \sum_{j=1}^{n_i} w_{ij}^2 \left\{ \left(\sum_{j=1}^{n_i} w_{ij} \right)^2 \right\}^{-1}$, where w_{ij} is the weight of household *j* in district *i*. Taking logarithms on both sides leads to

$$\log \text{GVF}_{i} = \alpha_{0}^{*} + \alpha_{1} \log \left[\hat{P}_{i} \left(1 - \hat{P}_{i} \right) \right] + \alpha_{2} \log \left(R_{i} \right).$$

This model can be fitted using the lm function in R to obtain the estimates $\hat{\alpha}_0^*$, $\hat{\alpha}_1$ and $\hat{\alpha}_2$ of regression coefficients. We then compute the smoothed GVF estimates of the sampling variance as $\hat{s}_i^2 = v^* (p_{iw}) = \exp(\hat{\alpha}_0^*) \times [\hat{P}_i(1-\hat{P}_i)]^{\hat{\alpha}_1} \times (R_i)^{\hat{\alpha}_2}$. Finally, for each district, we compute the effective sample size, $n_{i(e)}$ and the effective sample count of poor households, $y_{is(e)}$, as $n_{i(e)} = \frac{\hat{P}_i(1-\hat{P}_i)}{v^*(p_{iw})}$ and $y_{is(e)} = n_{i(e)}p_{iw}$. These values are rounded to the nearest integer. Figure 3.1 shows the effective sample sizes, plotted against the observed sample sizes. The effective sample counts and observed sample counts are shown in Figure 3.2. In the majority of cases the effective sample size is lower than the observed sample sizes. Similarly, in most of the cases, the effective sample counts are smaller than the observed sample counts. This indicates that the sampling design results in a loss in information, when compared with simple random sampling, in such districts. The district-wise survey weighted and unweighted direct estimates of proportion of poor households.

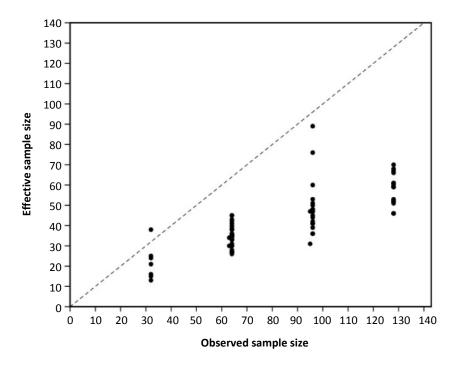


Figure 3.1 Effective sample size versus observed sample size for NSSO data.

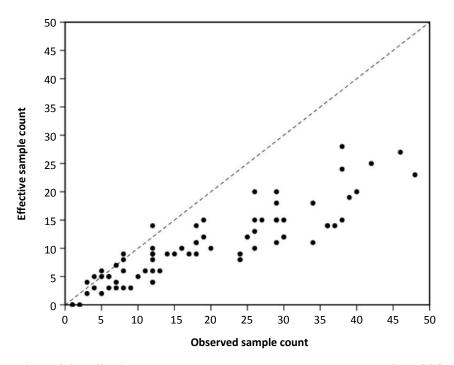


Figure 3.2 Effective sample count versus observed sample count for NSSO data.

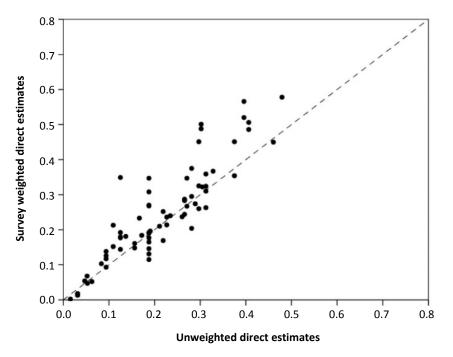


Figure 3.3 District-wise survey weighted direct estimates versus unweighted direct estimates of proportion of poor households.

3.3 Spatially correlated random area effects

The model (3.2) assumes that the random area effects are uncorrelated. In many applications the physical locations of the areas reflect missing contextual information and so this lack of correlation assumption for the area random effects is doubtful. It is often reasonable to assume that the effects of neighbouring areas (defined, for example, by a contiguity criterion) are correlated, with the correlation decaying to zero as the distance between these areas increases. In order to take into account the dependence between neighbouring areas, spatial models for random area effects are often used in SAE (Pratesi and Salvati, 2008; Chandra and Salvati, 2018). We achieve this by introducing spatial dependence in the error structure of model (3.2). In particular, we assume a Simultaneous Autoregressive (SAR) error process (Pratesi and Salvati, 2008), where the vector of random area effects $\mathbf{v} = (v_i)$ can be expressed as

$$\mathbf{v} = \rho \mathbf{L} \mathbf{v} + \mathbf{u},\tag{3.5}$$

where $\mathbf{v} = (\mathbf{I}_D - \rho \mathbf{L})^{-1} \mathbf{u}$, with $E(\mathbf{v}) = 0$ and $\operatorname{Var}(\mathbf{v}) = \sigma_u^2 [(\mathbf{I}_D - \rho \mathbf{L})(\mathbf{I}_D - \rho \mathbf{L}^T)]^{-1} = \Omega_{sp}(\delta)$. Here, **L** is a proximity matrix of order D, $\mathbf{u} \sim N(0, \sigma_u^2 \mathbf{I}_D)$ and ρ is a spatial autoregressive coefficient. From (3.2) and (3.5), the spatially dependent GLMM (SGLMM) is given by

$$\boldsymbol{\eta} = \mathbf{X}\boldsymbol{\beta} + \mathbf{Z}(\mathbf{I}_D - \boldsymbol{\rho}\mathbf{L})^{-1} \mathbf{u} = \mathbf{X}\boldsymbol{\beta} + \mathbf{Z}\mathbf{v}. \tag{3.6}$$

The proximity matrix **L** describes how random effects from neighbouring areas are related, whereas ρ defines the strength of this spatial relationship, i.e., the indicator of the degree to which one object is similar to other nearby objects. In what follows, we use the index "*sp*" to denote quantities associated with model (3.5). The simplest way to define **L** is as a contiguity matrix, i.e., **L** is a square binary matrix of order *D*, with non-zero values only for those pairs of areas that are adjacent. For ease of interpretation, this matrix is generally defined in row-standardized form; in which case ρ is called the spatial autocorrelation parameter (Banerjee, Carlin and Gelfand, 2004). Formally, the element l_{jk} (j, k = 1, ..., D) of a contiguity matrix takes the value 1 if area j shares an edge with area k and 0 otherwise. In row-standardised form this becomes

$$l_{jk} = \begin{cases} t_j^{-1} & \text{if } j \text{ and } k \text{ are contiguous} \\ 0 & \text{otherwise,} \end{cases}$$
(3.7)

where t_j is the total number of areas that share an edge with area j (including area j itself). Contiguity is arguably the simplest, but not necessarily the best, specification of a spatial interaction matrix, see for example Chandra (2013). It may be more informative to express this interaction in a more detailed way, e.g., as some function of the length of shared border between neighboring areas or as a function of the distances between the areas. In this paper, we therefore investigate a distance based definition for the proximity matrix L (i.e., a matrix with entries that are a function of the distances between the small areas or districts). In particular, we consider different ways of defining this proximity (or spatial weight matrix) with the aim of identifying the most effective approach to exploiting spatial information in order to produce reliable estimates for small areas. Let the spatial location of area or district *i* correspond to the coordinates of an arbitrarily defined spatial location (i.e., two dimensional x - y coordinates or latitude and longitude) in the area, e.g., its centroid, which we denote by $c_i = (c_{ix}, c_{iy})$. Let $d_{jk} = ||c_j - c_k||$ be an appropriate measure of the distance between the spatial locations of areas j and i. We then consider the following specifications for the proximity matrix **L** as function of distance:

(i) Proximity defined as the inverse of distance between the areas:

$$\mathbf{L} = \{l_{jk}\} = \begin{cases} d_{jk}^{-1} & j \neq k; \\ 0 & j = k. \end{cases}$$
(3.8)

(ii) An exponential specification for the proximity function, defined as

$$\mathbf{L} = \{l_{jk}\} = \exp\{-0.5d_{jk}^2\}$$
(3.9)

(iii) A Gaussian specification for the proximity function, defined as

$$\mathbf{L} = \{l_{jk}\} = \exp\{-0.5(d_{jk}/b)^2\}, \qquad (3.10)$$

where parameter b is the bandwidth, which can be optimally defined using a least squares criterion (Fotheringham, Brunsdon and Charlton, 2002). The bandwidth is a measure of how quickly the proximity decays with increasing distance. We use a cross validation procedure to estimate bandwidth. Here as the distance between areas j and k increases the proximity in (3.10) decreases exponentially. In particular, we use the *gwr.sel* function in the *spgwr* package of R to compute the value of the bandwidth. We also use the *Moran.I* function in the *ape* package of R to test for the presence of spatial correlation in the data. The results from this test show that there is evidence of spatial autocorrelation in the NSSO data. In particular, we reject the null hypothesis of zero spatial correlation at a 1 per cent level of significance.

In model (3.5), there are 2 parameters $(\sigma_u^2 \text{ and } \rho)$ that need to be estimated. Put $\mathbf{\delta} = (\sigma_u^2, \rho)^T$. Replacing these unknown parameters by their estimated values $\hat{\mathbf{\delta}} = (\hat{\sigma}_u^2, \hat{\rho})^T$, and denoting subsequent plug-in estimators by a "hat", we define the spatial empirical predictor (SEP) of the population count in area *i* as

$$\hat{y}_{i(e)}^{\text{SEP}} = y_{is(e)} + \left(N_i - n_{i(e)}\right) \hat{\pi}_{i(e)}^{\text{SEP}}, \qquad (3.11)$$

with $\hat{\pi}_{i(e)}^{\text{SEP}} = \exp i \left(\mathbf{x}_{i}^{T} \hat{\boldsymbol{\beta}}_{(e)}^{sp} + \mathbf{z}_{i}^{T} \hat{\mathbf{v}}_{(e)} \right)$. The corresponding spatial empirical predictor of the proportion in area *i* is $\hat{P}_{i(e)}^{\text{SEP}} = N_{i}^{-1} \hat{y}_{i(e)}^{\text{SEP}}$. For a non-sampled area, the spatial synthetic empirical predictor (hereafter denoted by SpSyn) of P_{i} is $\hat{P}_{i(e)}^{\text{SpSyn}} = \exp i \left(\mathbf{x}_{i,\text{out}}^{T} \hat{\boldsymbol{\beta}}_{(e)}^{sp} \right)$. Here $\hat{\boldsymbol{\beta}}_{(e)}^{sp}$ is the estimate of the fixed effect parameters $\boldsymbol{\beta}$ and $\hat{\mathbf{v}}_{(e)}$ is the predicted value of the spatially correlated random effects \mathbf{v} under the SAR model, using "effective sample size" and "effective sample counts". The estimation of unknown model parameters in (3.6) follows from the procedure discussed in previous section. However, the variance component σ_{u}^{2} is now $\boldsymbol{\delta} = (\sigma_{u}^{2}, \rho)^{T}$ and the predicted random effect $\hat{\mathbf{u}}_{(e)}$ is replaced by $\hat{\mathbf{v}}_{(e)}$.

3.4 Mean squared error (MSE) estimation

The MSE estimation of the small area empirical predictor (3.4) follows along the same lines as reported in Saei and Chambers (2003), Manteiga et al. (2007), Johnson et al. (2010), Chandra et al. (2011), Chandra, Salvati and Chambers (2018) and references therein. The estimate of the MSE of the empirical predictor (3.4) given by expression (3.12) is directly used, replacing the observed sample sizes and the observed sample counts by the effective sample sizes and the effective sample counts, respectively in order to incorporate the survey weights. Under model (3.2), using the effective sample sizes and the effective sample counts, an approximate estimate of the MSE of the EP (3.4) is

$$\operatorname{mse}\left(\hat{P}_{i(e)}^{\text{EP}}\right) \approx m_{1i}\left(\hat{\sigma}_{u}^{2}\right) + m_{2i}\left(\hat{\sigma}_{u}^{2}\right) + 2m_{3i}\left(\hat{\sigma}_{u}^{2}\right).$$
(3.12)

This estimate of MSE is based on an approximation that is analogous to the one used with the linear mixed model, see Rao and Molina (2015, Chapter 5, page 100-107), Saei and Chambers (2003) and Manteiga et al. (2007). To define three components of MSE (3.12), let $\hat{\Sigma} = \mathbf{Z}^T \mathbf{B} \mathbf{Z} + \hat{\mathbf{\Omega}}^{-1}$, $\mathbf{X}^* = \{ \operatorname{diag}(N_i^{-1}) \} \mathbf{H} \mathbf{X}$ and $\mathbf{Z}^* = \{ \operatorname{diag}(N_i^{-1}) \} \mathbf{H} \mathbf{Z}$, where

$$\mathbf{H} = \frac{\partial h(\mathbf{\eta})}{\partial \mathbf{\eta}} \bigg|_{\hat{\mathbf{\eta}}=\mathbf{\eta}} = \operatorname{diag} \left\{ \hat{P}_{i(e)}^{\mathrm{EP}} \left(1 - \hat{P}_{i(e)}^{\mathrm{EP}} \right); \ i = 1, \dots, D \right\}$$

and

$$\mathbf{B} = -\frac{\partial^2 l_1}{\partial \boldsymbol{\eta} \partial \boldsymbol{\eta}^T} \bigg|_{\hat{\boldsymbol{\eta}} = \boldsymbol{\eta}} = \operatorname{diag} \left\{ n_{i(e)} \hat{P}_{i(e)}^{\text{EP}} \left(1 - \hat{P}_{i(e)}^{\text{EP}} \right); \ i = 1, \dots, D \right\}$$

is the matrix of second derivatives of l_1 (the log-likelihood function l_1 defined by the vector $\mathbf{y}_{s(e)}$ given **u**) with respect to $\mathbf{\eta}$ at $\mathbf{\eta} = \hat{\mathbf{\eta}}$. Following McGilchrist (1994), we can write the variance-covariance matrix as

$$\hat{\mathbf{V}} = \begin{bmatrix} \mathbf{X}^T \\ \mathbf{Z}^T \end{bmatrix} \mathbf{B} \begin{bmatrix} \mathbf{X} & \mathbf{Z} \end{bmatrix} + \begin{bmatrix} \mathbf{0} & \mathbf{0} \\ \mathbf{0} & \hat{\mathbf{\Omega}}^{-1} \end{bmatrix} = \begin{pmatrix} \mathbf{X}^T \mathbf{B} \mathbf{X} & \mathbf{X}^T \mathbf{B} \mathbf{Z} \\ \mathbf{Z}^T \mathbf{B} \mathbf{X} & \hat{\mathbf{\Sigma}} \end{pmatrix} = \begin{pmatrix} \hat{\mathbf{V}}_{11} & \hat{\mathbf{V}}_{12} \\ \hat{\mathbf{V}}_{21} & \hat{\mathbf{V}}_{22} \end{pmatrix},$$

so that

$$\hat{\mathbf{V}}^{-1} = \begin{pmatrix} \mathbf{X}^T \mathbf{B} \mathbf{X} & \mathbf{X}^T \mathbf{B} \mathbf{Z} \\ \mathbf{Z}^T \mathbf{B} \mathbf{X} & \hat{\mathbf{\Sigma}} \end{pmatrix}^{-1} = \begin{pmatrix} \hat{\mathbf{A}}_{11} & \hat{\mathbf{A}}_{12} \\ \hat{\mathbf{A}}_{21} & \hat{\mathbf{A}}_{22} \end{pmatrix}$$

where we have partitioned both $\hat{\mathbf{V}}$ and its inverse $\hat{\mathbf{V}}^{-1}$ according to the dimensions of $\boldsymbol{\beta}$ and \mathbf{u} . Here $\hat{\mathbf{A}}_{11} = \begin{bmatrix} \mathbf{X}^T \mathbf{B} \mathbf{X} - \mathbf{X}^T \mathbf{B} \mathbf{Z} \hat{\boldsymbol{\Sigma}}^{-1} \mathbf{Z}^T \mathbf{B} \mathbf{X} \end{bmatrix}^{-1}$ and $\hat{\mathbf{A}}_{22} = \hat{\boldsymbol{\Sigma}}^{-1} + \hat{\boldsymbol{\Sigma}}^{-1} \{ (\mathbf{Z}^T \mathbf{B} \mathbf{X}) \hat{\mathbf{A}}_{11} (\mathbf{X}^T \mathbf{B} \mathbf{Z}) \} \hat{\boldsymbol{\Sigma}}^{-1}$. We put $\Delta = \mathbf{Z}^* \hat{\boldsymbol{\Sigma}}^{-1}$ and let $\mathbf{Z}_{(k)}^*$ denote the k^{th} row of the matrix \mathbf{Z}^* , with its derivative given by

$$\nabla_{(k)} = \frac{\partial \Delta_{(k)}}{\partial \sigma_u^2} \bigg|_{\sigma_u^2 = \hat{\sigma}_u^2} = \frac{\partial \left(\mathbf{Z}_{(k)}^* \mathbf{\Sigma}^{-1} \right)}{\partial \sigma_u^2} \bigg|_{\sigma_u^2 = \hat{\sigma}_u^2} = \left(\hat{\sigma}_u^2 \right)^{-2} \mathbf{Z}_{(k)}^* \hat{\mathbf{\Sigma}}^{-1} \hat{\mathbf{\Sigma}}^{-1} = \mathbf{Z}_{(k)}^* \hat{\mathbf{\Sigma}}^{-1} \hat{\mathbf{\Omega}}^{-1} \hat{\mathbf{\Omega}}^{-1} \hat{\mathbf{\Omega}}^{-1} \hat{\mathbf{\Sigma}}^{-1}$$

where $\hat{\Sigma}^+ = \mathbf{Z}^T (\mathbf{B} + \mathbf{B}\mathbf{Z}\hat{\Omega}\mathbf{Z}^T\mathbf{B})\mathbf{Z}$. With this notation, assuming model (3.2) holds and using the effective sample sizes and the effective sample counts, the components of MSE estimate (3.12) are

$$m_{1i} \left(\hat{\sigma}_{u}^{2} \right) = \mathbf{z}_{i}^{T} \left(\mathbf{Z}^{*} \hat{\mathbf{\Sigma}}^{-1} \mathbf{Z}^{*T} \right) \mathbf{z}_{i},$$

$$m_{2i} \left(\hat{\sigma}_{u}^{2} \right) = \mathbf{z}_{i}^{T} \left(\mathbf{C} \hat{\mathbf{A}}_{11} \mathbf{C}^{T} \right) \mathbf{z}_{i} \text{ with } \mathbf{C} = \mathbf{X}^{*} - \mathbf{Z}^{*} \hat{\mathbf{\Sigma}}^{-1} \mathbf{Z}^{T} \mathbf{B} \mathbf{X}, \text{ and}$$

$$m_{3i} \left(\hat{\sigma}_{u}^{2} \right) = \mathbf{z}_{i}^{T} \left\{ \text{trace} \left[\left(\nabla_{(k)} \hat{\mathbf{\Sigma}}^{*} \nabla_{(l)}^{T} \right) v \left(\hat{\sigma}_{u}^{2} \right) \right] \right\} \mathbf{z}_{i}.$$

Here $v(\hat{\sigma}_u^2)$ is the asymptotic covariance matrix of $\hat{\sigma}_u^2$, which is obtained as the inverse of the appropriate Fisher information matrix for $\hat{\sigma}_u^2$. If the REML estimate of σ_u^2 is used, then $v(\hat{\sigma}_u^2) = 2((\hat{\sigma}_u^2)^{-2}(D-2t_1)+(\hat{\sigma}_u^2)^{-4}t_{11})^{-1}$ with $t_1 = (\hat{\sigma}_u^2)^{-1}$ trace (\hat{A}_{22}) and $t_{11} = \text{trace}(\hat{A}_{22}\hat{A}_{22})$.

The MSE estimate of the SEP (3.11) is defined similarly. Replacing $\Omega(\hat{\sigma}_{u}^{2})$ by $\Omega_{sp}(\hat{\delta})$ and $\hat{\mathbf{u}}$ by $\hat{\mathbf{v}}$ in (3.12), this MSE estimate is

$$\operatorname{mse}\left(\hat{P}_{i(e)}^{\mathrm{SEP}}\right) = m_{1i}^{sp}\left(\hat{\boldsymbol{\delta}}\right) + m_{2i}^{sp}\left(\hat{\boldsymbol{\delta}}\right) + 2m_{3i}^{sp}\left(\hat{\boldsymbol{\delta}}\right), \qquad (3.13)$$

where under (3.5), the three components of (3.13) are defined as follows. Put

$$\hat{\boldsymbol{\Sigma}}_{sp} = \boldsymbol{Z}^T \boldsymbol{B}_{sp} \boldsymbol{Z} + \hat{\boldsymbol{\Omega}}_{sp}^{-1},$$

$$\boldsymbol{Z}_{sp}^* = \{ \operatorname{diag}(N_i^{-1}); \ i = 1, \dots, D \} \boldsymbol{H}_{sp} \boldsymbol{Z},$$

$$\boldsymbol{X}_{sp}^* = \{ \operatorname{diag}(N_i^{-1}); \ i = 1, \dots, D \} \boldsymbol{H}_{sp} \boldsymbol{X},$$

with

$$\mathbf{B}_{sp} = \operatorname{diag}\left\{n_{d}\hat{P}_{i(e)}^{\operatorname{SEP}}\left(1-\hat{P}_{i(e)}^{\operatorname{SEP}}\right); \ i=1, \ldots, D\right\}$$

and

$$\mathbf{H}_{sp} = \operatorname{diag}\left\{\hat{P}_{i(e)}^{\operatorname{SEP}}\left(1-\hat{P}_{i(e)}^{\operatorname{SEP}}\right); \ i=1,\ldots,D\right\}.$$

Then

$$m_{1i}^{sp}\left(\hat{\boldsymbol{\delta}}\right) = \mathbf{z}_{i}^{T}\left(\mathbf{Z}_{sp}^{*}\hat{\boldsymbol{\Sigma}}_{sp}^{-1}\mathbf{Z}_{sp}^{*T}\right)\mathbf{z}_{i},$$

$$m_{2i}^{sp}\left(\hat{\boldsymbol{\delta}}\right) = \mathbf{z}_{i}^{T}\left\{\mathbf{C}_{sp}\left(\mathbf{X}^{T}\mathbf{B}_{sp}\mathbf{X}-\mathbf{X}^{T}\mathbf{B}_{sp}\mathbf{Z}\hat{\boldsymbol{\Sigma}}_{sp}^{-1}\mathbf{Z}^{T}\mathbf{B}_{sp}\mathbf{X}\right)^{-1}\mathbf{C}_{sp}^{T}\right\}\mathbf{z}_{i}$$

with $\mathbf{C}_{sp} = \mathbf{X}_{sp}^* - \mathbf{Z}_{sp}^* \hat{\mathbf{\Sigma}}_{sp}^{-1} \mathbf{Z}^T \mathbf{B}_{sp} \mathbf{X}$, and

$$m_{3i}^{sp}\left(\hat{\boldsymbol{\delta}}\right) = \mathbf{z}_{i}^{T}\left\{\text{trace}\left[\left(\boldsymbol{\nabla}_{(k)}^{sp}\hat{\boldsymbol{\Sigma}}_{sp}^{+}\boldsymbol{\nabla}_{(l)}^{spT}\right)\boldsymbol{v}\left(\hat{\boldsymbol{\delta}}\right)\right]\right\}\mathbf{z}_{i}$$

with $\hat{\Sigma}_{sp}^+ = \mathbf{Z}^T \left(\mathbf{B}_{sp} + \mathbf{B}_{sp} \mathbf{Z} \hat{\Omega}_{sp} \mathbf{Z}^T \mathbf{B}_{sp} \right) \mathbf{Z}$. Here, $v\left(\hat{\delta} \right)$ is the asymptotic covariance matrix of the estimators of the variance component parameters $\hat{\delta}$,

$$\boldsymbol{\nabla}_{(k)}^{sp} = \frac{\partial \Delta_{(k)}^{sp}}{\partial \hat{\boldsymbol{\delta}}} \bigg|_{\hat{\boldsymbol{\delta}} = \hat{\boldsymbol{\delta}}} = \frac{\partial \left(\boldsymbol{Z}_{sp(k)}^{*} \boldsymbol{\Sigma}_{sp}^{-1} \right)}{\partial \left(\sigma_{u}^{2}, \rho \right)^{T}} \bigg|_{\hat{\boldsymbol{\delta}} = \hat{\boldsymbol{\delta}}} = \boldsymbol{Z}_{sp(k)}^{*} \hat{\boldsymbol{\Sigma}}_{sp}^{-1} \boldsymbol{\Omega}_{sp}^{-1} \left(\hat{\boldsymbol{\delta}} \right) \boldsymbol{\Omega}_{sp}^{-1} \left(\hat{\boldsymbol{\delta}} \right) \hat{\boldsymbol{\Sigma}}_{sp}^{-1},$$

where $\Delta^{sp} = \mathbf{Z}_{sp}^* \hat{\mathbf{\Sigma}}_{sp}^{-1}$ and $\mathbf{Z}_{sp(k)}^*$ is the k^{th} row of \mathbf{Z}_{sp}^* .

4 Empirical results

In this section we discuss application of the SAE methods introduced in the previous section to real survey data in order to generate estimates of the proportion of poor rural households at district level in the State of Uttar Pradesh in India. We use survey data from the 2011-12 HCES of the Indian NSSO and the 2011 Indian Population Census, and assume a binomial specification for the "effective" district level sample counts of poor rural households. The small area predictors that we consider, and their associated MSE estimators, are summarised in Table 4.1. We first describe some important diagnostic measures that can be used to examine the assumptions of the underlying models, and to validate the empirical performances of the different SAE methods.

Table 4.1Definition of various small area predictors

Predictor	Description	MSE Estimation
DIR	Direct survey estimate	Variance of DIR given in Section 3
EP	EP(3.4) under model (3.1)	MSE estimate (3.12)
SEP1	Spatial EP (3.11) under model (3.6) + weights (3.7)	MSE estimate (3.13)
SEP2	Spatial EP (3.11) under model (3.6) + weights (3.8)	MSE estimate (3.13)
SEP3	Spatial EP (3.11) under model (3.6) + weights (3.9)	MSE estimate (3.13)
SEP4	Spatial EP (3.11) under model (3.6) + weights (3.10)	MSE estimate (3.13)

4.1 Diagnostic measures

In SAE applications, two types of diagnostic measures are commonly employed, model diagnostics and diagnostics for the small area estimates (Brown, Chambers, Heady and Heasman, 2001). The main purpose of model diagnostics is to verify the distributional assumptions of the underlying small area model, i.e., how well this working model performs when it is fitted to the survey data. The small area estimate diagnostics, on the other hand, provide an indication of the reliability (and validity) of the model-based estimates produced by different SAE methods.

In the small area models defined by (3.1) and (3.6), the random area effects are assumed to have a normal distribution with mean zero and fixed variance. If the model assumptions are satisfied then the area (or district) level residuals are expected to be randomly distributed around zero. These residuals are calculated as $r_i = \hat{\eta}_{i(e)} - \mathbf{x}_i^T \hat{\boldsymbol{\beta}}_{(e)}$ and $r_i = \hat{\eta}_{i(e)} - \mathbf{x}_i^T \hat{\boldsymbol{\beta}}_{(e)}^{sp}$ under models (3.1) and (3.6), respectively. Histograms and normal probability (q - q) plots can be used to examine the normality assumption. Figure 4.1a, b, c displays the histogram of the district-level residuals (upper plot), the normal probability (q - q) plot of the district-level residuals (middle plot) and the distribution of the district-level residuals (lower plot) for different small area methods. We also use the Shapiro-Wilk test (via the *shapiro.test* function in R) to test the normality of the random area effects. These test results for the different SAE methods are reported in Table 4.2, with p -values lower than 0.05 indicating that the data deviate from normality. In Figure 4.1a, b, c, the district level residuals for all SAE methods appear to be randomly distributed around zero, as expected. In addition, the histograms and the q - q plots in Figure 4.1a, b, c also provide evidence in support of the normality assumption. Also, from Table 4.2, we see that the Shapiro-Wilk p -values are large for all the SAE methods. We conclude that the area random effects are likely to be normally distributed.

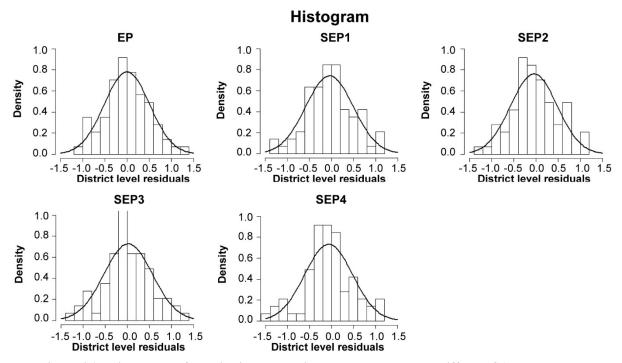


Figure 4.1a Histograms of the district-level residuals generated by the different SAE methods.

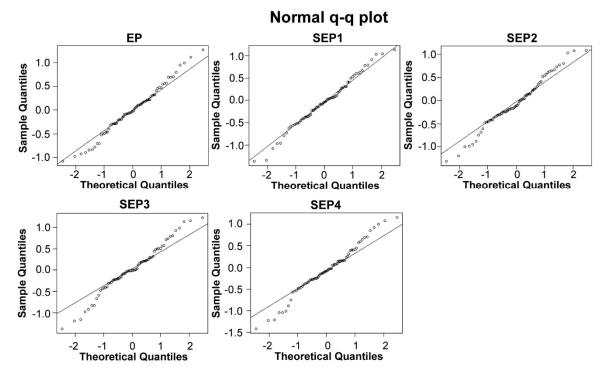


Figure 4.1b Normal q – q plots of the district-level residuals generated by the different SAE methods.

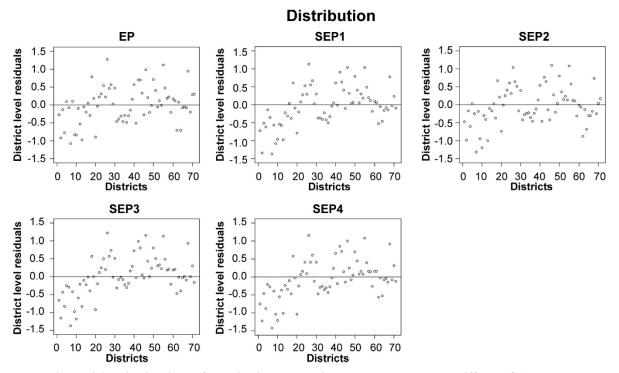


Figure 4.1c Distributions of the district-level residuals generated by the different SAE methods.

 Table 4.2

 Shapiro-Wilk test results for normality of district-level random effects

SAE Method	Test statistics value (W)	p-value
EP	0.988	0.764
SEP1	0.989	0.771
SEP2	0.984	0.491
SEP3	0.982	0.391
SEP4	0.982	0.399

A set of diagnostics that can be used for assessing the validity and the reliability of the model-based small area estimates are described in Brown et al. (2001). These diagnostics are based on the argument that model-based small area estimates should be (a) consistent with unbiased direct survey estimates, i.e., they should provide an approximation to the direct survey estimates that is consistent with these values being "close" to the expected values of the direct estimates; and (b) more precise than direct survey estimates, as evidenced by lower mean squared error estimates, i.e., the model-based small area estimates should have mean squared errors significantly lower than the variances of corresponding direct survey estimates. We consider four commonly used diagnostics measures that address these requirements, a bias diagnostic, a goodness of fit test, a 95 percent confidence interval diagnostic, and a percent coefficient of variation (CV) diagnostic. The first two diagnostics assess the validity and last two assess the reliability or improved precision of the model based small area estimates. In addition, we implemented a calibration diagnostic

where the model-based estimates are aggregated to higher level and compared with direct survey estimates at this level. See for example, Chandra et al. (2011). Note that here direct estimates are defined as the survey weighted direct estimates.

The bias diagnostic is based on following idea. The direct survey estimates are unbiased estimates of the population values of interest (i.e., true values). If the model-based estimates are "close" to the small area values of interest, then unbiased direct survey estimators should behave like random variables whose expected values correspond to the values of the model-based estimates. A Goodness of Fit (GoF) diagnostic, which is equivalent to a Wald test, for whether the differences between direct and model-based estimates have a zero mean, can therefore be applied (Brown et al., 2001). This Wald test statistic is computed as

$$W = \sum_{i} \frac{(DIR_{i} - Model-based estimate_{i})^{2}}{v(DIR_{i}) + mse(Model-based estimate_{i})}.$$

The value of W is compared against the 95th percentile of a chi square distribution with D degrees of freedom. Here D = 71 and this 95th percentile value is 91.670. The results from GoF diagnostic are set out in Table 4.3. We also calculated the average bias (Bias) and average relative difference (RE) between the direct and the model-based estimates, where

Bias =
$$D^{-1} \left(\sum_{i} \text{Model-based estimate}_{i} - \text{DIR}_{i} \right)$$

and

$$RE = D^{-1} \sum_{i} \left(\frac{Model-based estimate_{i} - DIR_{i}}{DIR_{i}} \right).$$

Table 4.4 shows the values of Bias and RE for different SAE methods. The results set out in Tables 4.3 and 4.4 clearly show that the model-based estimates generated by the different SAE methods are consistent with the direct survey estimates. Finally, in Figure 4.2 we provide a set of bias diagnostic plots, defined by plotting direct survey estimates (*Y*-axis) against corresponding model-based estimates (*X*-axis) and testing for divergence of the regression line from the Y = X line. These plots show that the model-based estimates are less extreme when compared to the direct survey estimates, demonstrating the typical SAE outcome of shrinking more extreme values towards the average. The values of R² for the fitted (OLS) regression line between the direct survey estimates and the EP, SEP1, SEP2, SEP3 and SEP4 estimates are 97, 93, 94, 92 and 96 per cent respectively. In general, these different bias diagnostics all show that the estimates generated by all the SAE methods appear to be consistent with the direct survey estimates.

Table 4.3Goodness of Fit Diagnostic

SAE Method	Goodness of Fit*	
EP	23.645	
SEP1	30.727	
SEP2	27.784	
SEP3	30.930	
SEP4	24.442	

*A small value (< 91.670 in this case) indicates no statistically significant difference between model-based and direct estimates.

8		
SAE Method	Bias	RE
EP	0.0023	0.2068
SEP1	0.0007	0.2155
SEP2	0.0016	0.1935
SEP3	0.0013	0.2096
SEP4	0.0016	0.2028

 Table 4.4

 Bias diagnostics between for direct survey (weighted) versus model based estimates

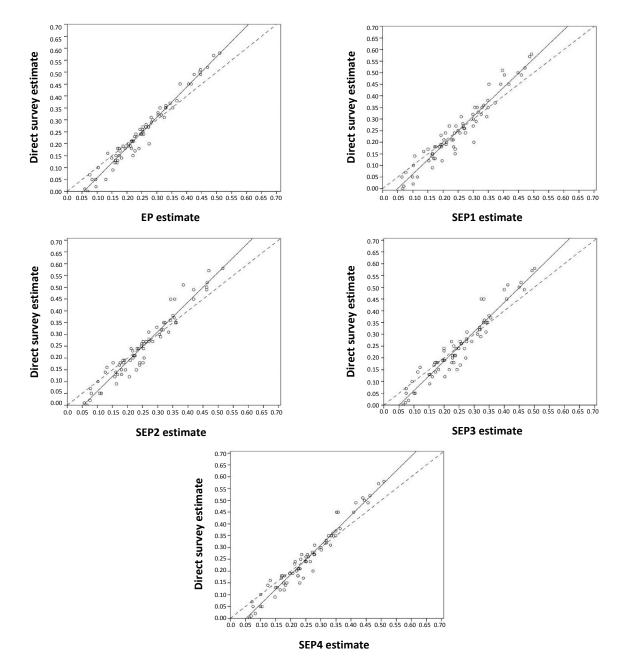


Figure 4.2 Bias diagnostic plots with y = x line (dotted) and regression line (solid) for proportion of poor rural households in Uttar Pradesh: model-based estimates versus direct survey (weighted) estimates.

A second set of diagnostics assess the reliability and improved precision of the model-based estimates relative to the direct survey estimates. The percent coefficient of variation (CV) is the estimated sampling standard error as a percentage of the estimate. Small area estimates with large CVs are considered unreliable. There is no international standard for what constitutes "too large" in this context (Chandra et al., 2011 and Johnson et al., 2010), but in general, CVs below 10 per cent are preferable for district level estimates. Table 4.5 provides a summary of CVs of the direct survey estimates and the model-based estimates generated by the different SAE methods. The results set out in Table 4.5 lead to two conclusions. First, the CVs of the direct estimates are larger than the CVs of the model-based estimates. Furthermore, the relative performances of the model-based methods as compared to the direct survey estimates improve with decreasing district specific observed (or effective) sample sizes. That is, the estimates computed from the different model-based approaches are more reliable and provide a better indication of rural poverty incidence in Uttar Pradesh. Second, among five model-based SAE methods considered in this analysis, the CVs of the spatial EP (3.11), based on the spatial model (3.6) with proximity matrix defined by the four different spatial proximity functions corresponding to SEP1, SEP2, SEP3 and SEP4 are all smaller than the CV of the EP (3.4) under the non-spatial model (3.1). That is, the use of spatial information improves the efficiency of the SAE method. In particular, SEP3 appears to be the best performing method of the four spatial methods that were investigated. In what follows we therefore focus on the SEP3 predictor only.

Table 4.5	
Distributions of CVs for different method	ds

Value	DIR	EP	SEP1	SEP2	SEP3	SEP4
Minimum	12.92	13.02	12.82	12.73	12.62	12.73
Q1	22.16	20.26	19.02	19.91	19.02	19.59
Median	30.81	24.42	23.76	24.20	22.92	23.78
Mean	35.68	25.35	24.11	24.94	23.95	24.80
Q3	47.64	30.52	27.78	29.12	27.97	29.36
Maximum	99.06	43.21	39.49	42.97	38.56	42.00

Figure 4.3 shows the district-wise distribution of the CVs of the DIR, EP and SEP3 methods. These show that direct survey estimates of poverty incidence are unstable with CVs that vary from 12.92 to 99.06% with average of 35.68%. Furthermore, the CVs of the direct survey estimates are greater than 40% and 50% in 7 and 14 (out of 71) districts respectively (see Figure 4.3). In contrast, the average CV values of EP and SEP3 are 25.35% and 23.95% respectively, and the CV of SEP3 is smaller than that of EP in 69 out of 71 districts.

The district-wise plot of the 95 percent confidence intervals (CIs) generated by DIR and SEP3 are displayed in Figure 4.4. This shows that the 95% CIs for the direct estimates are wider than the 95% CIs for the model based estimates generated by SEP3. We supplement this visual exploration of 95% CIs by computing the values of the coverage diagnostic described in Brown et al. (2001) as a way of evaluating the validity of the confidence intervals generated by the model-based estimates. Here we first calculate adjusted 95% confidence intervals for the direct and model-based estimates using the district-wise critical values

 $c_i = 1.96 \left\{ \sqrt{v(\text{DIR}_i) + \text{mse}(\text{model-based estimate}_i)} / \left(\sqrt{v(\text{DIR}_i)} + \sqrt{\text{mse}(\text{model-based estimate}_i)} \right) \right\}.$

We then count the number of times these adjusted intervals do not overlap. Nominally, this non-coverage rate should be at most 5%. The observed non-coverage rates for EP and SEP3 methods are 1.41% and 2.82% respectively.

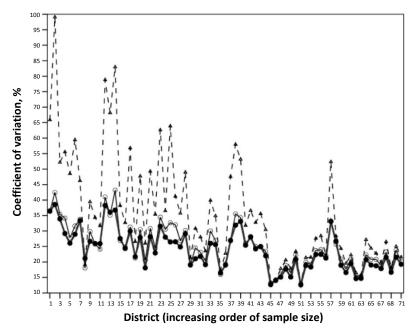


Figure 4.3 District-wise coefficients of variation (%) plot for the DIR (dash line, Δ), EP (thick line,°) and SEP3 (solid line,•) estimates.

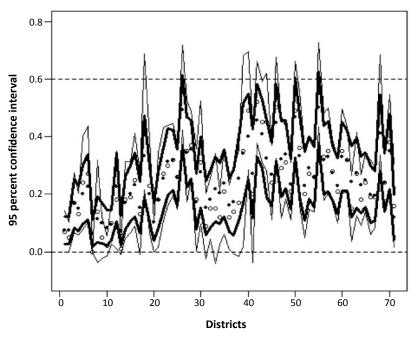


Figure 4.4 District-wise 95 percent confidence interval plots for the direct estimates (thick line, °) versus SEP3 estimates (solid line,•).

Finally, we investigate the calibration properties of the model-based district-level estimates at higher (e.g., State or Region) level. Let \hat{P}_i and N_i denote the estimate of proportion of poor household and population size for district *i*. The state-level estimate of the proportion of poor households is calculated as $\hat{P} = \sum_{i=1}^{D} N_i \hat{P}_i / \sum_{i=1}^{D} N_i$. Uttar Pradesh is divided into Central, Eastern, Western and Southern regions, and calibration properties can also be examined for these regions. State and regional level estimates of the proportion of poor rural households generated by the different SAE methods are reported in Table 4.6. Comparing these with the corresponding direct estimates we see that the model-based estimates are very close to the direct survey estimates as state level as well in each of the four regions.

Table 4.6

Aggregated level estimates of proportion of poor household generated by different SAE methods. These estimates are aggregated over 71 districts at state level as well as four regional levels

Region	DIR	EP	SEP1	SEP2	SEP3	SEP4
State	0.26	0.26	0.26	0.26	0.26	0.26
Central	0.34	0.32	0.34	0.33	0.33	0.33
Eastern	0.31	0.30	0.31	0.30	0.31	0.31
Southern	0.26	0.27	0.27	0.27	0.27	0.28
Western	0.15	0.17	0.16	0.17	0.16	0.16

4.2 Discussion of results

The analysis set out in the previous sub-section clearly demonstrates that model-based estimates generated by the different methods considered in the study are consistent with the direct survey estimates. Furthermore, CV values, 95% CIs and coverage diagnostic values also show that the model-based estimates are reliable and more stable than the corresponding direct survey estimates. Consequently the model-based SAE estimates, and in particular those generated by the SEP3 method, can be recommended for use by various stakeholders for policy planning and implementation. In Figure 4.5 we display the poverty map showing the estimated proportion of poor rural households (i.e., poverty incidence) in different districts of Uttar Pradesh produced by the SEP3 method. This map provides poverty distribution in the state of Uttar Pradesh. This map shows the district-wise degree of inequality with respect to the distribution of poor rural households, defined as a household having monthly per capita consumer expenditure below the state poverty line. An accompanying map showing the CV of SEP3 in the different districts is also presented in Figure 4.6. These maps are supplemented by the results set out in Table 4.7, where we report the district-wise estimates along with CVs and 95% confidence intervals generated by DIR and SEP3. For example, in the western part of Uttar Pradesh there are many districts with low rural poverty incidence such as Saharanpur, Hathras, Meerut, Baghpat, Muzaffarnagar, Bulandshahar etc. Similarly, in the eastern part and in Bundelkhand region (north-east part of map) we see a number of districts (for example, Azamgarh, Sitapur, Chitrakoot, Bahraich, Siddharthnagar, Banda, Fatehpur, Basti and Kaushambi etc) with a high level of rural poverty incidence. Further, in Table 4.6 we see that the SEP estimated rural poverty incidence is lowest for the Western region (16%) and highest for the Central (33%) region, while the state average is around 26%. In Table 4.7, the

district-wise estimates of rural poverty incidence generated by SEP3 vary from minimum of 6% to a maximum of 50%. The spatial inequality in the distribution of rural poverty incidence in different districts of the State of Uttar Pradesh is clearly visible in Figure 4.5 and from the estimates set out in Tables 4.6 and 4.7. This should prove useful for policy planners and administrators aiming to take effective financial and administrative decisions.

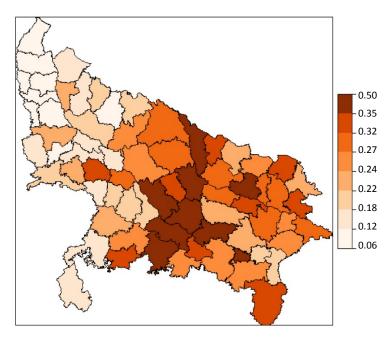


Figure 4.5 District-wise distribution of rural poverty incidence in the State of Uttar Pradesh generated by SEP3 method.

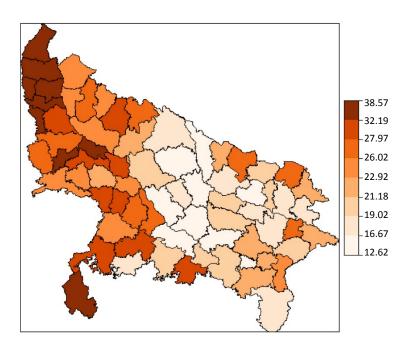


Figure 4.6 District-wise distribution of percent coefficient of variation for SEP3 method.

Table 4.7

District-wise direct estimates (DIR) and model-based estimates (SEP3) along with their CVs (expressed in percentage terms) and corresponding 95 percent confidence intervals for poverty incidence in the state of Uttar Pradesh in 2011-12

Region	District	Estimated poverty incidence		95 percent confidence intervals				Coefficient of variation, %	
		DIR	SEP3	D	IR	SE	2P3	DIR	SEP3
				Lower	Upper	Lower	Upper		
Western	Saharanpur	0.07	0.07	0.00	0.14	0.03	0.12	53.16	33.10
	Muzaffarnagar	0.05	0.07	0.00	0.10	0.03	0.12	52.36	33.10
	Bijnor	0.17	0.17	0.06	0.28	0.08	0.25	31.79	25.41
	Moradabad	0.13	0.15	0.06	0.20	0.07	0.23	28.41	26.67
	Rampur	0.24	0.20	0.08	0.40	0.10	0.30	34.35	25.98
	Jyotiba P. Nagar	0.27	0.23	0.10	0.44	0.11	0.34	31.69	25.92
	Meerut	0.00	0.06	0.00	0.00	0.02	0.11	78.83	38.27
	Baghpat	0.14	0.11	-0.04	0.32	0.03	0.19	65.87	36.49
	Ghaziabad	0.05	0.10	-0.02	0.12	0.03	0.17	68.16	36.06
	G. B. Nagar	0.02	0.08	-0.02	0.06	0.02	0.14	99.06	38.56
	Bulandshahr	0.10	0.09	0.03	0.17	0.04	0.15	36.60	28.15
	Aligarh	0.18	0.23	0.01	0.35	0.11	0.35	47.56	26.91
	Mahamaya Nr	0.01	0.07	-0.01	0.03	0.02	0.12	82.97	36.75
	Mathura	0.18	0.17	0.05	0.31	0.08	0.26	38.23	27.59
	Agra	0.19	0.20	0.07	0.31	0.11	0.30	32.75	24.25
	Firozabad	0.25	0.23	0.09	0.41	0.12	0.34	32.69	24.38
	Etah	0.13	0.15	-0.01	0.27	0.06	0.24	56.82	29.95
	Mainpuri	0.45	0.33	0.22	0.68	0.19	0.47	26.57	21.72
	Budaun	0.23	0.20	0.07	0.39	0.10	0.30	35.57	25.00
	Bareilly	0.05	0.10	-0.01	0.11	0.04	0.17	57.98	31.89
	Pilibhit	0.18	0.18	0.01	0.35	0.08	0.28	47.72	27.93
	Shahjahanpur	0.27	0.25	0.11	0.43	0.14	0.36	30.39	22.01
	Farrukhabad	0.18	0.17	0.01	0.35	0.08	0.27	49.26	28.16
	Kannauj	0.31	0.28	0.10	0.52	0.15	0.40	35.37	22.92
	Etawah	0.09	0.15	-0.02	0.20	0.06	0.25	62.63	31.55
	Auraiya	0.15	0.18	0.04	0.26	0.08	0.28	36.53	28.02
	Kansiram Nr	0.16	0.12	0.02	0.30	0.04	0.20	46.20	33.33
Central	Kheri	0.30	0.31	0.16	0.44	0.20	0.43	24.31	18.96
	Sitapur	0.32	0.32	0.20	0.44	0.22	0.42	19.56	16.59
	Hardoi	0.26	0.26	0.15	0.37	0.16	0.36	22.25	19.46
	Unnao	0.57	0.49	0.42	0.72	0.37	0.61	13.36	12.69
	Lucknow	0.35	0.35	0.17	0.53	0.22	0.47	26.02	18.10
	Rae Bareli	0.37	0.36	0.25	0.49	0.25	0.46	16.46	14.64
	Kanpur Dehat	0.15	0.22	-0.04	0.34	0.10	0.33	63.88	26.47
	Kanpur Nagar	0.12	0.20	0.02	0.22	0.10	0.31	41.19	26.53
	Fatehpur	0.52	0.46	0.38	0.66	0.33	0.58	13.88	14.04
Southern	Jalaun	0.21	0.24	0.06	0.36	0.12	0.35	35.70	24.86
	Jhansi	0.12	0.16	0.00	0.24	0.07	0.25	48.97	29.13
	Lalitpur	0.12	0.17	0.00	0.28	0.06	0.29	52.16	33.90
	Hamirpur	0.14	0.17	-0.02	0.26	0.00	0.29	55.55	29.20
	Banda	0.17	0.23	0.28	0.30	0.11	0.40	21.36	29.20 19.04
	Chitrakoot	0.49	0.40	-0.03	0.70	0.23	0.33	59.47	28.95
	Mahoba	0.35	0.33	0.20	0.50	0.21	0.45	21.38	18.87

Table 4.7 (continued)

District-wise direct estimates (DIR) and model-based estimates (SEP3) along with their CVs (expressed in percentage terms) and corresponding 95 percent confidence intervals for poverty incidence in the state of Uttar Pradesh in 2011-12

Region	District	Estimated pov	verty incidence	95 percent confidence intervals				Coefficient of variation, %	
		DIR	SEP3	DIR		SE	P3	DIR	SEP3
				Lower	Upper	Lower	Upper		
Eastern	Maharajganj	0.35	0.34	0.02	0.68	0.17	0.52	48.51	26.07
	Pratapgarh	0.45	0.41	0.31	0.59	0.29	0.53	15.77	14.71
	Kaushambi	0.45	0.32	0.28	0.62	0.19	0.46	19.16	21.16
	Allahabad	0.24	0.27	0.11	0.37	0.16	0.38	27.21	20.78
	Bara Banki	0.50	0.45	0.32	0.68	0.32	0.58	17.86	15.11
	Faizabad	0.29	0.32	0.11	0.47	0.19	0.45	30.81	21.06
	Ambedkar Nr	0.31	0.34	0.18	0.44	0.22	0.46	20.62	17.70
	Sultanpur	0.21	0.24	0.11	0.31	0.15	0.32	24.39	19.03
	Bahraich	0.49	0.47	0.32	0.66	0.33	0.61	17.90	15.14
	Shrawasti	0.36	0.33	0.16	0.56	0.19	0.48	28.08	21.80
	Balrampur	0.20	0.23	0.05	0.35	0.11	0.35	39.42	26.68
	Gonda	0.27	0.29	0.15	0.39	0.18	0.40	22.81	18.76
	Siddharth Nr	0.26	0.26	0.14	0.38	0.15	0.36	23.43	20.87
	Basti	0.58	0.50	0.43	0.73	0.38	0.62	12.92	12.62
	Sant Kabir Nr	0.33	0.32	0.18	0.48	0.20	0.44	23.54	19.13
	Gorakhpur	0.28	0.27	0.17	0.39	0.18	0.37	19.90	17.88
	Kushinagar	0.21	0.23	0.10	0.32	0.13	0.32	26.53	21.49
	Deoria	0.35	0.33	0.20	0.50	0.21	0.45	21.49	18.38
	Azamgarh	0.32	0.32	0.21	0.43	0.21	0.42	18.15	16.69
	Mau	0.15	0.24	0.03	0.27	0.12	0.37	39.89	26.03
	Ballia	0.27	0.28	0.12	0.42	0.15	0.40	27.64	22.42
	Jaunpur	0.18	0.23	0.09	0.27	0.13	0.33	25.01	21.55
	Ghazipur	0.24	0.25	0.14	0.34	0.16	0.34	21.56	19.26
	Chandauli	0.19	0.20	0.06	0.32	0.10	0.30	34.87	25.62
	Varanasi	0.19	0.19	0.08	0.30	0.11	0.28	28.43	22.47
	S. R. Das Nr	0.51	0.41	0.34	0.68	0.28	0.54	17.40	16.46
	Mirzapur	0.24	0.24	0.14	0.34	0.14	0.34	22.06	21.25
	Sonbhadra	0.38	0.35	0.21	0.55	0.22	0.48	22.77	21.25

5 Concluding remarks

In this article we describe a spatial extension of the area level version of the GLMM and consider SAE of survey weighted proportions under this model. We introduce spatial dependence in the error structure of GLMM through a SAR process, and we explore different ways of incorporating the resulting spatial dependence into the small area estimates. We use effective sample size and effective sample count to account for the sampling design used in the survey. An analytical MSE estimator is also described. We then apply these SAE methods to real survey data in order to estimate rural poverty incidence and to produce a corresponding poverty map of the different districts for the Indian state of Uttar Pradesh. We evaluate our empirical results using several diagnostic measures and show that the model-based SAE methods provide

significant gains in efficiency for generating district level estimates of proportions of poor households. Furthermore, when spatial correlation between the districts is used appropriately, the model-based estimates lead to significant gains in terms of efficiency for the district level estimates. For these data, a spatial proximity measure defined as a Gaussian function of the distances between districts appears to be the most effective way of incorporating spatial association between the districts. Finally, use of survey information through effective sample size leads to more representative and realistic estimates.

In the context of estimation of small area counts under unit level small area models, D'Alò, Di Consiglio, Falorsi, Ranalli and Solari (2012) describe several different ways that spatial information using different distance measures can be included in analysis. An exploration of these alternative approaches to including spatial information under area level versions of the corresponding small area models therefore seems worthwhile. Furthermore, in this article we use a SAR structure to capture the spatial dependence in the GLMM. However, as noted earlier, there are alternative approaches such as the use of CAR models as well as newly developed models for spatial nonstationarity that can be used to characterise this spatial dependence. See, for example, Besag et al. (1991), Leroux et al. (1999), Chandra et al. (2017, 2018). These approaches should be explored, and we are currently carrying out further research on them.

Finally, we note that the district level estimates of rural poverty incidence produced using the methods outlined in this paper will be useful for various Departments and Ministries in Government of India as well as International organizations for their policy research and strategic planning. They will also be useful for budget allocation and to target welfare interventions by identifying the districts/regions with high rural poverty incidence. This application also provides evidence that SAE can be used as cost effective and efficient approach for generating reliable disaggregate level statistics from existing survey data by combining auxiliary information from different published sources with direct survey estimates.

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Abstract

Small area estimation using area-level models can sometimes benefit from covariates that are observed subject to random errors, such as covariates that are themselves estimates drawn from another survey. Given estimates of the variances of these measurement (sampling) errors for each small area, one can account for the uncertainty in such covariates using measurement error models (e.g., Ybarra and Lohr, 2008). Two types of area-level measurement error models have been examined in the small area estimation literature. The functional measurement error model assumes that the underlying true values of the covariates with measurement error are fixed but unknown quantities. The structural measurement error model assumes that these true values follow a model, leading to a multivariate model for the covariates observed with error and the original dependent variable. We compare and contrast these two models with the alternative of simply ignoring measurement error when it is present (naïve model), exploring the consequences for prediction mean squared errors of use of an incorrect model under different underlying assumptions about the true model. Comparisons done using analytic formulas for the mean squared errors assuming model parameters are known yield some surprising results. We also illustrate results with a model fitted to data from the U.S. Census Bureau's Small Area Income and Poverty Estimates (SAIPE) Program.

Key Words: Sample survey; Area level model; Covariate; Prediction.

1 Introduction

Linear mixed models, particularly that of Fay and Herriot (1979), have gotten great attention in small area estimation. The Fay-Herriot (FH) model can be written

$$Y_i = \theta_i + e_i \qquad \theta_i = z'_i \delta + u_i \qquad i = 1, \dots, m \tag{1.1}$$

where, for areas *i* indexed from 1 to *m*, the Y_i are direct survey estimates of population quantities θ_i , the sampling errors e_i in Y_i are assumed independent $N(0, D_i)$ with the D_i taken as known (they are actually estimated using survey micro-data), the z_i are $q \times 1$ vectors of regression covariates with corresponding coefficient vector δ , and the random effects u_i are distributed i.i.d. $N(0, \sigma_u^2)$ and independently of the e_i .

In some cases it may be desired to augment the model for θ_i with one or more covariates X_i that are themselves estimates taken from another survey that estimates characteristics believed to be related to θ_i . One approach is to simply ignore the sampling error in X_i , treating it like the covariates in z_i which we shall assume are not subject to sampling or other measurement errors. We shall call this the *naïve Fay-Herriot model*, which, taking for simplicity the case of a single such covariate X_i , we write as

$$Y_i = \theta_i + e_i \qquad \theta_i = \beta_N X_i + z'_i \delta_N + u_{i,N}. \tag{1.2}$$

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We add the "N" subscripts to the regression coefficients and the random effects $(u_{i,N})$ to distinguish this model from the measurement error models to come. The model assumes that $u_{i,N} \sim i.i.d. N(0, \sigma_{u,N}^2)$, although with heteroscedastic sampling error in X_i the assumption that $var(u_{i,N})$ is constant is incorrect, implying that the model (1.2) is misspecified. This point is discussed further below.

An alternative to the naïve FH model is to use a measurement error model to account for the sampling (measurement) error in X_i . Assume x_i denotes the population characteristic being estimated by X_i with sampling error η_i , where the η_i are assumed distributed independently $N(0, C_i)$ and with the C_i taken as known (actually estimated using survey micro-data). A generalization of the model (1.1) to include the covariate X_i while accounting for its sampling error is

$$Y_i = \theta_i + e_i \qquad \theta_i = \beta x_i + z_i' \delta + u_i \tag{1.3}$$

$$X_i = x_i + \eta_i. \tag{1.4}$$

If the x_i are assumed to be fixed unknown quantities, then the model defined by (1.3)-(1.4) is known as the *functional measurement error model* (FME model). This model is discussed by Fuller (1987) and has been studied for small area estimation by Ybarra and Lohr (2008), Arima, Datta and Liseo (2015, 2016), and Arima, Bell, Datta, Franco and Liseo (2017). Analogous unit level measurement error models for small area estimation have been studied by Ghosh and Sinha (2007), Datta, Rao and Torabi (2010), and Arima, Datta, and Liseo (2012).

Another alternative to the naïve FH model is to specify a model for x_i in (1.4) which, with (1.3), implies bivariate models for $(\theta_i, x_i)'$ and $(Y_i, X_i)'$. This is known as a *structural measurement error model* (SME model). If x_i follows the regression model $x_i = z'_{xi}\delta_x + v_i$, with covariates z_{xi} and residuals $v_i \sim i.i.d$. $N(0, \sigma_v^2)$ independent of u_i , then the resulting model for $(Y_i, X_i)'$ can be written as

$$\begin{bmatrix} Y_i \\ X_i \end{bmatrix} = \begin{bmatrix} \theta_i \\ \eta_i \end{bmatrix} + \begin{bmatrix} e_i \\ \eta_i \end{bmatrix} \sim \text{ i.i.d. } N(0, \Omega) \qquad \Omega = \begin{bmatrix} D_i & 0 \\ 0 & C_i \end{bmatrix}$$
(1.5)

$$= \left(\begin{bmatrix} z'_i & \beta z'_{xi} \\ 0 & z'_{xi} \end{bmatrix} \begin{bmatrix} \delta \\ \delta_x \end{bmatrix} + \begin{bmatrix} u_i + \beta v_i \\ v_i \end{bmatrix} \right) + \begin{bmatrix} e_i \\ \eta_i \end{bmatrix}$$
(1.6)

$$\begin{bmatrix} u_i + \beta v_i \\ v_i \end{bmatrix} \sim \text{ i.i.d. } N(0, \Sigma) \qquad \Sigma = \begin{bmatrix} \sigma_u^2 + \beta^2 \sigma_v^2 & \beta \sigma_v^2 \\ & & \\ \beta \sigma_v^2 & \sigma_v^2 \end{bmatrix}.$$
(1.7)

This model differs from a standard bivariate FH model in that the parameter β affects both the regression mean function for Y_i and the random effect covariance matrix Σ . However, if the covariates z_{xi} are linear functions of the covariates z_i , then the fixed effects regression part of (1.6) can be reparameterized to unrestricted linear regression effects $[z'_i \delta_y z'_{xi} \delta_x]'$ with regression covariates z_i for the first equation and z_{xi} for the second. With this reparameterization β no longer affects the regression fixed effects, so the matrix Σ can then be reparameterized in the general form $\Sigma = [\sigma_{jk}]$, or by σ_{11}, σ_{22} , and

 $\rho = \sigma_{12} / \sqrt{\sigma_{11}\sigma_{22}} \in [-1, 1]$, as there is now a 1-1 correspondence between $(\sigma_u^2, \sigma_v^2, \beta)$ and $(\sigma_{11}, \sigma_{22}, \sigma_{12})$ or $(\sigma_{11}, \sigma_{22}, \rho)$. Two instances where this condition on z_{xi} holds are (i) if the regression covariates are the same in both equations $(z_{xi} = z_i)$, or (ii) if z_{xi} is just an intercept term $(z_{xi} = 1)$ and z_i also includes an intercept.

Datta, Delaigle, Hall and Wang (2018) study the area level SME model, while Huang and Bell (2012) present a study examining use of general bivariate models for small area estimation. Analogous unit level models have been studied by Ghosh, Sinha and Kim (2006) and Torabi, Datta and Rao (2009). Fuller (1987) and Buonaccorsi (2010) discuss additional measurement error models including nonlinear models and the Berkson model.

Note that the FME and SME models model the relation between the true unobserved quantities θ_i and x_i , whereas the naïve FH model models the relation between θ_i and the observed X_i . The X_i contain noise in the form of generally heteroscedastic sampling error, and this heteroscedasticity produces the naïve model's misspecification noted earlier.

If Y_i and X_i are estimates from the same survey sample their sampling errors e_i and η_i are likely to be correlated. This can be accommodated by replacing the off-diagonal 0 of Ω in (1.5) by the appropriate $\operatorname{cov}(e_i, \eta_i)$ (estimated using survey micro-data). While this works for the SME model, correlation between e_i and η_i implies that the regressor X_i and sampling error e_i are correlated, violating a fundamental assumption of the FH model and causing potentially severe problems for the naïve FH model. Hence, we do not consider that situation here.

In this paper we compare the three alternative models – naïve FH, functional measurement error, and structural measurement error – focusing on their predictive performance for small area estimation. One motivating case involves the use of the naïve FH model when measurement error (η_i) is present, comparing the naïve FH model's predictive accuracy with those of the other two models. We also compare the predictive performance of the functional versus structural measurement error models. We make these comparisons using analytic formulas for the mean squared errors (MSEs) for the case where model parameters are known (first order approximations). This provides good approximations for the case when the number of areas *m* is large. It is also relevant as the typically dominant term in the MSEs for smaller values of *m*. Since the naïve FH model is misspecified, we make precise in what sense its parameters are "known".

Section 2 summarizes some theoretical results for the three alternative models, first on convergence of parameter estimates and then on small area prediction, covering both the point predictors and their MSEs. We provide results for the three models first for the case where the FME model is true, and then for the case where the SME model is true. Derivations of these results are deferred to a corresponding technical report (Bell, Chung, Datta, and Franco, 2018). Section 3 compares, via contour plots, the theoretical MSEs of small area predictors for the three models across ranges of the parameters of a true SME model. Section 4 uses the theoretical MSE formulas to compare prediction MSEs from the three models when they are applied to

an empirical example of modeling poverty rates of school-age children for U.S. counties. The example is taken from the U.S. Census Bureau's Small Area Income and Poverty Estimates (SAIPE) program. Section 5 then gives general conclusions.

2 Theoretical results

To facilitate interpretation of the results, here we use the simplest possible versions of the models outlined in the Introduction, specifically, models where the vector of non-measurement error covariates reduces to just an intercept term, i.e., $z_i = 1$. To revert to fairly standard notation, we use α for the intercept coefficient instead of δ , so the simplified model for θ_i in the FME and SME models (from (1.3)) becomes

$$\theta_i = \alpha + \beta x_i + u_i. \tag{2.1}$$

For the SME model we assume that $x_i \sim i.i.d. N(\mu, \sigma_x^2)$ so there are no regression terms other than the mean μ in the model for x_i .

For the naïve FH model (1.2), the simplified model for θ_i becomes

$$\theta_i = \alpha_N + \beta_N X_i + u_{i,N} \tag{2.2}$$

where, as before, we use the "N" subscript to distinguish the coefficients and random effects in the naïve model (2.2) for θ_i , since this model differs from (2.1) by substituting X_i in place of x_i .

We now give some results on parameter estimation and small area prediction for these models. The Appendix of Bell et al. (2018) provides derivations of these results.

2.1 Parameter estimators and their large sample limits

The Appendix of Bell et al. (2018) details unbiased estimating equations for the parameters of the three models considered here. The resulting estimators of α , β , and σ_u^2 for the simplified versions of the FME and SME models are the same, and are given by

$$\hat{\beta} = \frac{\frac{1}{m} \sum_{i=1}^{m} (X_i - \overline{X}) (Y_i - \overline{Y})}{\frac{1}{m} \sum_{i=1}^{m} (X_i - \overline{X})^2 - \overline{C}}$$

$$\hat{\alpha} = \overline{Y} - \hat{\beta} \overline{X}$$

$$\hat{\sigma}_u^2 = \frac{1}{m} \sum_{i=1}^{m} (Y_i - \hat{\alpha} - \hat{\beta} X_i)^2 - \overline{D} - \hat{\beta}^2 \overline{C}$$
(2.3)

where $\overline{X} = m^{-1} \sum_{i=1}^{m} X_i$, with analogous definitions of \overline{Y} , \overline{C} , and \overline{D} . For fitting the SME model, we also have $\hat{\mu} = \overline{X}$ and $\hat{\sigma}_x^2 = \frac{1}{m} \sum_{i=1}^{m} (X_i - \overline{X})^2 - \overline{C}$. Result 1 gives the probability limits of all these parameter estimators.

Result 1: For the FME and SME models given by (1.3)-(1.4) and by (1.5)-(1.7), respectively, but with the simplified model for θ_i as in (2.1), the parameter estimators given in (2.3) are consistent for the true model parameters, that is,

$$\hat{\beta} \xrightarrow{P} \beta \qquad \hat{\alpha} \xrightarrow{P} \alpha \qquad \hat{\sigma}_{u}^{2} \xrightarrow{P} \sigma_{u}^{2}$$
(2.4)

where $\stackrel{P}{\rightarrow}$ denotes convergence in probability as $m \rightarrow \infty$ under the true model (whether FME or SME). For fitting the SME model when it is true, we also have $\hat{\mu} \stackrel{P}{\rightarrow} \mu$ and $\hat{\sigma}_x^2 \stackrel{P}{\rightarrow} \sigma_x^2$. For fitting the SME model when the FME model is true, we have $\hat{\mu} \stackrel{P}{\rightarrow} \overline{x} = \lim_{m \rightarrow \infty} \frac{1}{m} \sum_{i=1}^{m} x_i$ and $\hat{\sigma}_x^2 \stackrel{P}{\rightarrow} s_x^2 = \lim_{m \rightarrow \infty} \frac{1}{m} \sum_{i=1}^{m} (x_i - \overline{x})^2$, with both limits assumed to exist.

Remark 1: The estimators in (2.3) are the same for the two models despite being obtained from different estimating equations – see equations (19) and (20) versus equations (36)-(38) in Bell et al. (2018). The consistency results in (2.4) thus hold whether the true model is the FME or the SME. These consistency results also hold for the more general versions of these models considered in the Appendix of Bell et al. (2018).

The parameter estimators for fitting the naïve FH model with the simplified model for θ_i as in (2.2) are given by

$$\hat{\beta}_{N} = \frac{\frac{1}{m} \sum_{i=1}^{m} (X_{i} - \bar{X}) (Y_{i} - \bar{Y})}{\frac{1}{m} \sum_{i=1}^{m} (X_{i} - \bar{X})^{2}} = \frac{\frac{1}{m} \sum_{i=1}^{m} (X_{i} - \bar{X})^{2} - \bar{C}}{\frac{1}{m} \sum_{i=1}^{m} (X_{i} - \bar{X})^{2}} \times \hat{\beta}$$

$$\hat{\alpha}_{N} = \bar{Y} - \hat{\beta}_{N} \bar{X}$$

$$\hat{\sigma}_{u,N}^{2} = \frac{1}{m} \sum_{i=1}^{m} (Y_{i} - \hat{\alpha}_{N} - \hat{\beta}_{N} X_{i})^{2} - \bar{D}.$$
(2.5)

Result 2: When the FME model (or SME model) is true, the parameter estimators in (2.5) have the following probability limits:

$$\hat{\beta}_{N} \xrightarrow{P} a\beta \quad \hat{\alpha}_{N} \xrightarrow{P} \alpha + (1-a)\beta\overline{x} \quad \hat{\sigma}_{u,N}^{2} \xrightarrow{P} \sigma_{u}^{2} + a\beta^{2}\overline{C} \quad (\text{FME model true})$$
$$\hat{\beta}_{N} \xrightarrow{P} a_{*}\beta \quad \hat{\alpha}_{N} \xrightarrow{P} \alpha + (1-a_{*})\beta\mu \quad \hat{\sigma}_{u,N}^{2} \xrightarrow{P} \sigma_{u}^{2} + a_{*}\beta^{2}\overline{C} \quad (\text{SME model true})$$

where the "attenuation factors" are given by

$$a = \frac{s_x^2}{s_x^2 + \overline{C}} \quad \text{(FME model true)} \quad a_* = \frac{\sigma_x^2}{\sigma_x^2 + \overline{C}} \quad \text{(SME model true)}. \tag{2.6}$$

Remark 2: The results for convergence of $\hat{\beta}_N$ in Result 2 are versions of the well-known attenuation of the estimate of the regression parameter when measurement error is ignored – see Theil (1971, page 608)

for the FME case and Fuller (1987, page 3) for the SME case. The limit for $\hat{\sigma}_{u,N}^2$ shows that the naïve FH model inflates the estimate of the model error variance σ_u^2 for the true model (whether FME or SME) by the amount $a\beta^2\overline{C}$ (or $a_*\beta^2\overline{C}$) due to the failure of the naïve model to account for the measurement error.

2.2 Small area predictors and their MSEs

Result 3 lists the formulas for the predictors of θ_i for the three models. Note that any of these formulas will apply whenever the corresponding model is assumed and used for prediction, regardless of whether the true model is the FME, the SME, or some other model. The FME predictor $\hat{\theta}_{i,F}$ is given by Theorem 1 of Ybarra and Lohr (2008), while the naïve FH predictor $\hat{\theta}_{i,N}$ is simply the predictor of Fay and Herriot (1979) for the case of our simplified naïve FH model. Derivations for the more general models are given in the Appendix of Bell et al. (2018).

Result 3: The predictors of θ_i from the simple versions of the FME, SME, and naïve FH models considered here are as follows:

FME predictor:
$$\hat{\theta}_{i,F} = Y_i - \frac{D_i \left\{Y_i - \hat{\alpha} - \hat{\beta}X_i\right\}}{D_i + \hat{\sigma}_u^2 + \hat{\beta}^2 C_i}$$

SME predictor: $\hat{\theta}_{i,S} = Y_i - \frac{D_i \left\{Y_i - \hat{\alpha} - \hat{\beta}X_i + \hat{\beta}\left(C_i / (\hat{\sigma}_x^2 + C_i)\right)(X_i - \overline{X})\right\}}{D_i + \hat{\sigma}_u^2 + \hat{\beta}^2 C_i \hat{\sigma}_x^2 / (\hat{\sigma}_x^2 + C_i)}$
naïve FH predictor: $\hat{\theta}_{i,N} = Y_i - \frac{D_i \left\{Y_i - \hat{\alpha}_N - \hat{\beta}_N X_i\right\}}{D_i + \hat{\sigma}_{u,N}^2}.$

These are the empirical versions of the optimal predictors (best linear unbiased predictors) under their respective assumed models.

Remark 3: Several special cases are worth noting from these results. First, as $D_i \rightarrow 0$ all the predictors converge to the direct survey estimate Y_i , and since its sampling variance is then going to 0, all the predictors achieve design consistency assuming that Y_i is itself design consistent. Second, if $C_i = \overline{C}$ it can be shown that the SME and naïve FH predictors agree while the FME predictor generally remains different. (The Appendix of Bell et al. (2018) shows that, for the more general model considered there, the SME and naïve FH predictors agree asymptotically when $C_i = \overline{C}$.) Third, it can be seen that as $\hat{\sigma}_x^2 \rightarrow \infty$ the SME predictor converges to the FME predictor. The same holds as $C_i \rightarrow 0$, which implies in the limit that x_i is known. We can put these together and say that the SME and FME predictors behave similarly when C_i / σ_x^2 is small.

Remark 4: It can be shown that the formula for $\hat{\theta}_{i,s}$ can be obtained by taking the formula for $\hat{\theta}_{i,F}$ and replacing X_i in the numerator of the fraction by $E(x_i | X_i) = X_i - E(\eta_i | x_i) = X_i - [C_i / (\hat{\sigma}_x^2 + C_i)](X_i - \overline{X})$, and $C_i = \operatorname{var}(X_i - x_i)$ in the denominator of the fraction by $\operatorname{var}(x_i | X_i) = C_i \hat{\sigma}_x^2 / (\hat{\sigma}_x^2 + C_i)$, these being the conditional mean and variance of x_i given X_i under the estimated model.

Table 2.1 gives, for the case when the FME model is true, the first order biases and prediction error variances of the three predictors. The MSEs are then the squared biases plus the variances. (The prediction error variance, and thus the MSE, for the FME model is given by Theorem 1 of Ybarra and Lohr (2008).) The results assume the true FME model parameters are known, but for the naïve FH model they account for the fact that the estimates of the parameters are biased as shown in Result 2. This gives a realistic approximation for the case when m, the number of areas, is large. The Table 2.1 entries for the SME model use the quantity

$$F_{i} = (\sigma_{u}^{2} + D_{i})(s_{x}^{2} + C_{i}) + \beta^{2}s_{x}^{2}C_{i}.$$
(2.7)

 Table 2.1

 Biases and prediction error variances when the FME model is true

Prediction model	Bias	Prediction error variance
FME	0	$\frac{(\sigma_u^2 + \beta^2 C_i)D_i}{\sigma_u^2 + \beta^2 C_i + D_i}$
SME	$\frac{-\beta D_i C_i}{F_i} (x_i - \overline{x})$	$D_{i} - \frac{D_{i}^{2} \left[\left(s_{x}^{2} + C_{i} \right) + \beta^{2} s_{x}^{2} C_{i}^{2} / F_{i} \right]}{F_{i}}$
naïve FH	$\frac{-\beta D_i (1-a)}{\sigma_u^2 + a\beta^2 \overline{C} + D_i} (x_i - \overline{x})$	$\frac{\left(\sigma_{u}^{2}+a\beta^{2}\overline{C}\right)D_{i}}{\sigma_{u}^{2}+a\beta^{2}\overline{C}+D_{i}}+\left(\frac{\beta D_{i}}{\sigma_{u}^{2}+a\beta^{2}\overline{C}+D_{i}}\right)^{2}a\left(aC_{i}-\overline{C}\right)$

Table 2.2 gives the results for the case when the SME model is true. In this case all the predictors are unbiased in the sense that $E(\hat{\theta}_i - \theta_i) = 0$. Hence, the table just gives the prediction error variances, which are also the MSEs. For F_i^* in Table 2.2, we substitute σ_x^2 for s_x^2 in the expression (2.7), analogous to the definition of a_* in (2.6).

Table 2.2Prediction error variances when the SME model is true

Prediction model	Prediction error variance = MSE
FME	$\frac{(\sigma_u^2 + \beta^2 C_i)D_i}{\sigma_u^2 + \beta^2 C_i + D_i}$
SME	$D_i - rac{D_i^2 \left(\sigma_x^2 + C_i ight)}{F_i^*}$
naïve FH	$\frac{\left(\sigma_u^2 + a_*\beta^2\overline{C}\right)D_i}{\sigma_u^2 + a_*\beta^2\overline{C} + D_i} + \left(\frac{\beta D_i}{\sigma_u^2 + a_*\beta^2\overline{C} + D_i}\right)^2 a_*^2 \left(C_i - \overline{C}\right)$

Several points are worth noting about the results of Tables 2.1 and 2.2.

1. The results for the FME predictor are the same in both cases, i.e., whether the FME or SME model is true. To achieve unbiasedness under the assumption that the x_i are fixed, unknown quantities, the FME predictor eliminates them from the prediction error. Hence, its prediction

error results are not affected by whether the x_i actually are fixed and unknown or are random variables following some distribution, as the SME model assumes.

- 2. When the FME model is true, the biases of the SME and naïve FH predictors are proportional to $(x_i \overline{x})$, which is unconstrained, and so can be arbitrarily large in magnitude. Hence, for areas where $|x_i \overline{x}|$ is large the squared bias can dominate the prediction MSE. Since the x_i are unobserved it will typically be difficult to estimate the squared bias (unless the C_i are small so the X_i are very good estimators of the x_i , in which case the motivation to use a measurement error model diminishes).
- 3. The MSEs in Table 2.2 for the SME and naïve models can be obtained by taking the expressions for squared bias plus prediction error variance from Table 2.1, substituting σ_x^2 for s_x^2 and also for $(x_i \overline{x})^2$, and simplifying. This is the difference between assuming the x_i fixed and unknown versus assuming $x_i \sim i.i.d. N(\mu, \sigma_x^2)$.
- 4. As noted earlier, if an area has $C_i = \overline{C}$ then the SME and naïve FH predictors agree. Hence, when $C_i = \overline{C}$ the biases, error variances, and MSEs of the SME and naïve FH predictors are the same. If the SME model is true then the SME predictor is optimal, and thus so is the naïve FH predictor for areas with $C_i = \overline{C}$, in which case both are superior to the FME predictor. In fact, comparing the MSEs of the naïve and FME predictors from Table 2.2, and given that $0 \le a_* < 1$, one can show directly that the FME predictor's MSE is larger when $C_i = \overline{C}$.
- 5. When the SME model is true, for areas with $C_i = \overline{C}$ the "reported MSE" for the naïve FH model will agree with the true MSE. The reported MSE is the MSE one would compute assuming the naïve model to be true, and is given by the first term in the naïve FH MSE expression in Table 2.2. The second term is obviously zero when $C_i = \overline{C}$, and is positive when $C_i > \overline{C}$. We thus see that when $C_i > \overline{C}$ the reported MSE understates the true MSE, while when $C_i < \overline{C}$ the second term in the MSE is negative so the reported MSE overstates the true MSE. The misspecification of the naïve FH model when the SME model is true can thus lead to substantial misstatement of the MSEs except for areas for which C_i is close to \overline{C} .

An implication of points 4 and 5, and the analogous result stated earlier for the point predictors, is that if $C_i = \overline{C}$ for all i = 1, ..., m, then the prediction results for the SME and naïve FH models are the same. This provides some basis for the statement sometimes made that measurement error in covariates doesn't affect model prediction. Put another way, this statement is true only if the C_i are constant for all areas, and only when comparing prediction results for the naïve FH model to those for the SME model. Prediction results for the FME model will be different.

3 Comparing MSEs for the alternative predictors

We now compare the performance of the three alternative model predictors when applied to data from a true SME model, making such comparisons across a range of values for the model parameters and the D_i

and C_i values. The comparisons use the MSE results of Table 2.2, examining percentage differences in the MSEs calculated, for example, as

$$100\left(\frac{\text{MSE}_{F}}{\text{MSE}_{S}} - 1\right)$$
(3.1)

for comparing MSEs of the FME and SME predictors. We similarly define the analogs to (3.1) for comparing MSEs for the naïve FH and FME predictors, and of the naïve FH and SME predictors, as well as for comparing the reported and actual MSEs of the naïve FH predictor. Assuming that the SME model is true facilitates the comparisons. Assuming that the FME model is true leads to the complication that the MSEs for the structural and naïve models depend on x_i (Table 2.1), which has unrestricted variation over areas. Section 4 nonetheless makes some MSE comparisons under a true FME model.

For making relative comparisons as in (3.1) the scale of the data doesn't matter, so rescaling to Y_i/σ_u will not affect these comparisons. This is also true for rescaling X_i to X_i/σ_x . These rescalings reduce the number of varying parameters we need to consider by two, which lets us express $1/\sigma_u^2$ times MSE_F, MSE_S, MSE_N, and \widehat{MSE}_N (the reported MSE for the naïve FH predictor), all computed assuming the SME model is true, in terms of the following four scale independent quantities:

$$r_{D} = \frac{D_{i}}{\sigma_{u}^{2}}, \quad r_{C} = \frac{C_{i}}{\sigma_{x}^{2}}, \quad \rho = \operatorname{corr}(\theta_{i}, x_{i}), \quad \overline{r}_{C} = \frac{\overline{C}}{\sigma_{x}^{2}}.$$
(3.2)

The Appendix illustrates such re-expression for the calculation of MSE_s / σ_u^2 . To simplify the notation, we omit the *i* subscript from r_D and r_C , though except in the unusual situation where the D_i and C_i are actually constant over areas, one needs to compute the MSE expressions separately for each area *i*. To compare MSEs we examine contour plots over (r_D, r_C) for each of the MSE percentage differences defined as in (3.1), viewing the MSE percentage difference as a function of (r_D, r_C) . We examine such plots for fixed values of ρ and $\overline{r_c}$ (which do not vary over *i*).

Figure 3.1 gives contour plots of (3.1) for $\rho = 0.3$ and $\rho = 0.7$. The x- and y-axes of the plots, representing the values of r_D and r_C , range from 0.1 to 10, and are shown with log scaling. We need not set \overline{r}_C for these comparisons because MSE_F and MSE_s do not depend on \overline{C} . The results in the plots are easy to summarize: the percentage differences are all positive, favoring the SME model which here is assumed to be true, and the differences increase with both r_D and r_C , so that the more sampling or measurement error is present, the larger is the advantage to use of the SME predictor. When either r_D or r_C is small, say generally below 1, the MSE percentage differences are small, which is why no contours show up plotted in this area, and choice of model has little effect on prediction accuracy. In fact, when both r_D and r_C are sufficiently small the FME and SME predictors are both close to the direct estimator Y_i , leading to small MSE differences, a pattern repeated in subsequent graphs. Towards the upper right corner the MSE percentage differences become substantial in both graphs, larger for $\rho = 0.7$. Analysis of the formula for MSE_F/MSE_s reveals that, for given values of r_C and r_D , the MSE percent differences increase with

 $\rho > 0$ to the point where $\rho = \left[1 + r_c / \sqrt{(1 + r_c)(1 + r_p)}\right]^{-0.5}$, and then they decline to 0 as ρ increases to 1. Over the range of values [1, 10] for r_c and r_p , this maximum point varies from about $\rho = 0.57$ to $\rho = 0.91$. Note that the results for ρ and for $-\rho$ would be the same since the MSEs actually depend on ρ^2 .

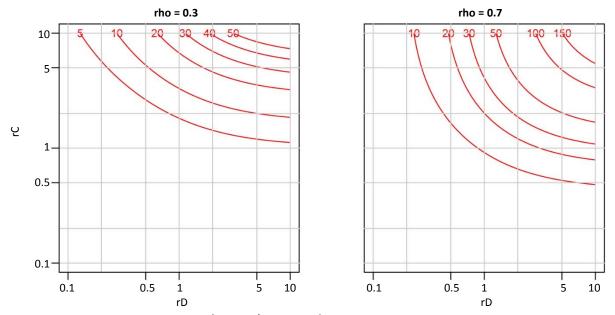


Figure 3.1 Contours of $100 (MSE_F / MSE_S - 1)$ for two values of ρ when the SME model is true.

For the other MSE comparisons, which are shown in Figures 3.2-3.4, the percentage differences depend on all four quantities in (3.2). To get a general idea of how the comparisons vary, we take $\rho = 0.7$ as a representative value, and examine contour plots for $\overline{r}_c = 0.1$, 1, and 10. It is worth noting that the analogous plots done for $\rho = 0.3$, 0.5, and 0.9, not shown here, present similar patterns, though with the patterns generally shifted somewhat in location on the plots, and typically with contours representing lower or higher percentage differences.

Figure 3.2 shows contour plots of $100 (MSE_N/MSE_F - 1)$, comparing MSEs for the naïve and FME predictors. In these plots we see both positive and negative contours, indicating regions where the FME predictor does better, and other parts where the naïve FH predictor does better. The patterns in these plots can be understood by keeping in mind that (i) for small values of r_c the FME predictor acts like the SME predictor, which here is optimal, so the FME predictor performs well, and (ii) for r_c close to $\overline{r_c}$ the naïve FH predictor acts like the SME predictor and so performs well. Thus, in the plot for $\overline{r_c} = 0.1$, both the FME and naïve FH predictors perform similarly to the optimal SME predictor for small values of r_c , so there is little difference in their MSEs. Apart from this case where both perform well, the naïve FH predictor performs better when r_c is sufficiently close to $\overline{r_c}$, where the meaning of "sufficiently close" depends on the values of $\overline{r_c}$ and r_p .

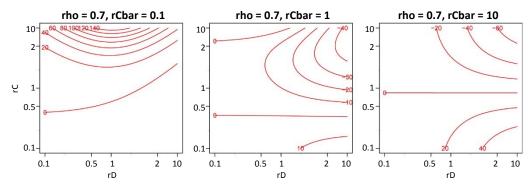


Figure 3.2 Contours of $100 (MSE_N / MSE_F - 1)$ for $\rho = 0.7$ and $\overline{r}_c = 0.1, 1, \text{ and } 10$ when the SME model is true.

The results in Figure 3.2 showing that for certain regions the naïve FH predictor has lower prediction MSE than the FME predictor may at first seem surprising given that, when the SME model is true, the naïve model is misspecified since it ignores the measurement error in X_i . In contrast, the FME model accounts for the measurement error in X_i and, since it makes no assumptions about the x_i , it is not inconsistent with the true SME model. In fact, as we move towards larger amounts of measurement error overall (larger values of \overline{r}_c), the MSE advantages of the naïve FH predictor become more substantial and cover larger ranges of the r_c and r_D values. The general explanation for this is that, when measurement error is substantial, the FME model's avoidance of any modeling assumptions about the x_i can lead to rather inefficient use of the data X_i , while the naïve FH predictor makes suboptimal but better use of the X_i unless r_c is very different from \overline{r}_c).

Figure 3.3 gives contour plots of $100 (MSE_N/MSE_S - 1)$, comparing MSEs for the naïve and SME predictors. Since the SME model is assumed true for the purposes of these computations, all the contours shown are positive, with the exception of a zero line in each plot (represented here by the contour plotting function of R (R core team, 2016) as a set of "0" labels not joined by a line). These zero contours occur as horizontal lines for $\overline{r_c} = 0.1$, 1, and 10 on the three plots, these being where $r_c = \overline{r_c}$, implying $C_i = \overline{C}$, which is when the naïve FH and SME predictors agree. Apart from this, the plots for $\overline{r_c} = 0.1$ and 1 show substantial positive contours for large values of r_c that also increase with r_D , while for $\overline{r_c} = 10$ the substantial positive contours occur for small r_c as r_D grows large.

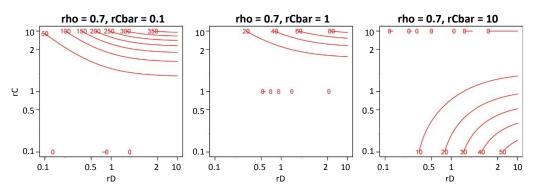


Figure 3.3 Contours of $100 (MSE_N / MSE_S - 1)$ for $\rho = 0.7$ and $\overline{r}_c = 0.1$, 1, and 10 when the SME model is true.

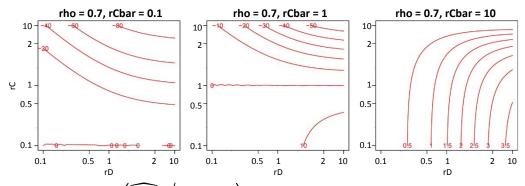


Figure 3.4 Contours of $100\left(\widehat{MSE}_N / MSE_N - 1\right)$ for $\rho = 0.7$ and $\overline{r}_c = 0.1, 1, \text{ and } 10$ when the SME model is true.

Figure 3.4 gives contour plots of $100 (\widehat{\text{MSE}}_N / \text{MSE}_N - 1)$, comparing the reported and actual MSEs for the naïve FH predictor. As with Figure 3.3, the three plots should show zero contours at the values $\overline{r}_c = 0.1$, 1, and 10, respectively (which are poorly represented in the first two plots, and absent from the third). In these plots the regions above the zero contours have negative values that reflect understatement of the true MSE by the reported MSE, while the regions below the zero contours have positive values that reflect overstatement of the true MSE. The first two plots show regions for $r_c > \overline{r}_c$ with significant understatement of the true MSE, while the second two reflect at most very minor overstatement of the true MSE when $r_c < \overline{r}_c$. This pattern remains when the axis ranges are expanded to include larger values of r_D and r_c . While further extrapolation of these results to more general cases than those considered here is questionable, they nonetheless suggest that understatement of MSE by the naïve FH model may be a potentially more serious problem than overstatement.

4 SAIPE illustration

The previous section compared the performance of the three alternative model predictors across a range of values for the model parameters and the D_i and C_i values for a true SME model. Here we take a model developed for an important small area application – modeling county poverty rates of school-age children for U.S. counties – to determine realistic values of the model parameters and the D_i and C_i values. We take the fitted model as a true model, and then use the theoretical formulas from Section 2 to compare small area prediction MSEs for the three alternative model predictors. We do this using the fitted SME model as truth, and then repeat the exercise using the corresponding FME model as truth. For the latter we simulate the true covariate values x_i from the fitted SME model, since the prediction biases and MSEs depend on these values which are not observed. We emphasize that our objective here is not in producing county poverty estimates; we use the poverty rate data merely to get a realistic model for illustrating the results from Section 2.

We fit the SME model to estimates of poverty rates of school-age children for U.S. counties from the American Community Survey, or ACS (U.S. Census Bureau, 2014), the largest U.S. household survey. ACS

produces annual estimates based on one year or five years of data collection. Here, we use 2010 ACS oneyear estimates of county poverty rates of school-age children as the primary response variable Y_i . We center the analogous 2005-2009 ACS five-year estimates, and treat them as a covariate X_i which is subject to measurement error. We also use covariates from administrative records as the covariates z_i not subject to measurement error. These are drawn from two sources – tabulations of income tax records obtained under an agreement with the U.S. Internal Revenue Service, as well as recipient counts from the Supplemental Nutrition Assistance Program, a program that provides food subsidies to low income households. The specific covariates used are the same as those of Arima et al. (2017), though that paper jointly modeled two years of poverty rates using a multivariate FME model. All covariates used here are centered about their means. The model we use here is similar to models applied to such data by Bell, Basel, Cruse, Dalzell, Maples, O'Hara and Powers (2007), and is related to the county production model used by the SAIPE Program. SAIPE produces poverty estimates at the state, county, and school district level for different age groups, including the school-age group 5-17. For more information about SAIPE, see Bell, Basel and Maples (2016) or the SAIPE web page at https://www.census.gov/programs-surveys/saipe.html/.

We fitted the SME model to the poverty rate data via maximum likelihood using R (R core team, 2016) to obtain values of the parameters defining our "true model". This yielded $\hat{\sigma}_u^2 = 0.0012$, $\hat{\sigma}_x^2 = 0.0064$, and $\hat{\beta} = 0.407$. These parameter values imply that $\hat{\rho} = \hat{\beta} \left[\hat{\sigma}_x^2 / (\hat{\sigma}_u^2 + \hat{\beta}^2 \hat{\sigma}_x^2) \right]^{0.5}$ (see Appendix) is about 0.7. We omit the estimates of the other model parameters since they do not affect the first order MSE calculations done here.

For the D_i and C_i values we used estimates from a Generalized Variance Function (GVF, see Wolter, 1985) developed for the sampling variances of the 2010 one-year and 2005-2009 five-year ACS county school-age poverty rate estimates, respectively. The specifics of the GVF are described in Franco and Bell (2013). After the SME model fitting, but for use when computing the MSEs, the D_i and C_i values were altered to protect against their disclosure by adding zero mean bivariate normal noise to the $\log(D_i)$ and $\log(C_i)$ values, and exponentiating the results. The noise terms added to the D_i and C_i had a correlation of 0.5 and variances of $2/n_i$ and $2/5n_i$, respectively, where the n_i are the 2010 one-year ACS county sample sizes. Thus, more noise was added to the $\log(D_i)$ than to the $\log(C_i)$, and more noise was added for counties with smaller sample sizes. The resulting D_i values range from about 0.00005 to 0.12 with a median of 0.0046, while C_i ranges from 7×10^{-6} to 0.013 with a median of 0.0009. Resulting values of the ratios $r_D = D_i/\hat{\sigma}_u^2$ range from about 0.04 to 100 with a median of about 4, and values of $r_C = C_i/\hat{\sigma}_x^2$ range from about 0.001 to 2 with a median of about 0.14. The noise altered values still provide a practically plausible range of values for the D_i and C_i , and the general appearance of the plots that follow was not materially changed by the noise infusion.

Figure 4.1 panels (a)-(c) display ratios comparing first order approximations of the three model predictors' MSEs plotted against C_i on the log scale, with a vertical line at $\overline{C} = 0.0014$ shown for reference. Panel (a) shows the ratios of MSEs for the SME and naïve predictors. We note that, due to their optimality under the assumed SME model, the SME model predictors always have lower prediction MSEs

than the naïve model predictors. Because the C_i 's are strongly related to the D_i 's (with a correlation of about 0.9), for small C_i 's the D_i 's are also likely to be small, and all three model predictors are then approximately equal to the direct estimators, so that the MSEs of the naïve and SME predictors are similar. We will see this trend in all four panels of Figure 4.1. In panel (a), the ratio reaches its maximum of approximately one when $C_i \approx \overline{C}$. This agrees with a result given in Remark 3 of Section 2.2, where we noted that the two predictors are equivalent at this point. For $C_i > \overline{C}$ the ratios decline rapidly to values approaching 30% larger MSEs for the naïve predictors.

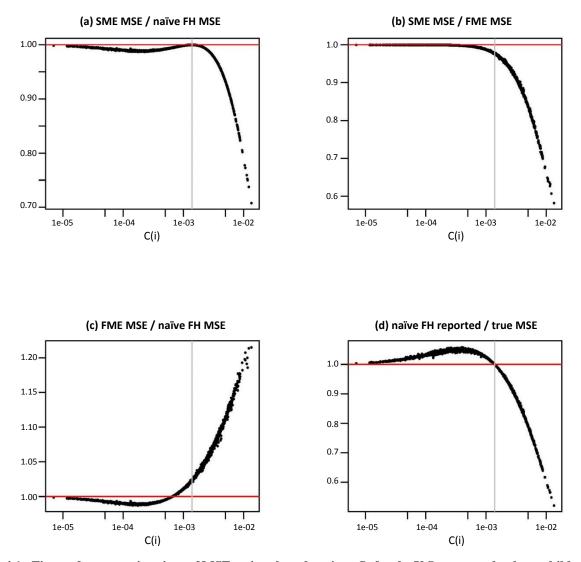


Figure 4.1 First order approximations of MSE ratios plotted against C_i for the U.S. county school-age children in poverty model when the SME model is true. For panels (a)-(c), the ratios are of the true MSEs of the SME and naïve, SME and FME, and FME and naïve models, respectively. Panel (d) shows the ratios of the reported MSEs and the true MSEs of the naïve model. The vertical lines mark \overline{C} .

Panel (b) shows the ratios of the SME and FME predictors' MSEs. Again, the SME predictor performs best since we are assuming the SME model is true. For $C_i < \overline{C}$ the MSE differences are small, but the differences become pronounced for high values of C_i , with the FME predictor MSEs up to 40% or more higher than those for the SME predictor.

Panel (c) shows the corresponding MSE ratios for the FME and naïve FH predictors. The naïve predictor has slightly higher MSEs than the FME predictor for small C_i but lower MSEs for large C_i 's, a pattern expected from the results in panels (a) and (b). The two predictors' MSEs are approximately equal at some value of $C_i < \overline{C}$. The FME predictor's MSEs are larger by about 20% or more than the naïve predictor's MSEs for the largest C_i values.

Note that the MSE that is obtained for the FME predictor when the SME model is actually true is still correct to the first order, though the FME predictor is not optimal. However, the MSE obtained assuming the naïve model is true, what we call the "reported" MSE, differs from the naïve model predictor's true MSE. Panel (d) plots the ratios of the first order approximations of the reported and true MSEs of the naïve model predictor when the SME model is true. As noted in Section 2.2, the naïve model overstates the MSEs for small C_i 's and understates them for large C_i 's, while correctly estimating the MSE at $C_i = \overline{C}$. The overstatement for $C_i < \overline{C}$ is relatively small, less than 10%, while the understatement for $C_i > \overline{C}$ becomes large, increasing with increasing C_i to more than 40%.

One might argue that the SME model is more reasonable than the FME model for this application, because if one is willing to assume a model for the true poverty rates as measured by the ACS 2010 estimates, why not assume a model for the true five-year average poverty rates as essentially measured by the ACS 2005–2009 estimates? Still, it is of interest to investigate the performance of each of the predictors when the FME model holds. This presents a further challenge because the true x_i 's are not known. For this illustration, we generate them as $x_i \stackrel{\text{iid}}{\sim} N(0, \hat{\sigma}_x^2)$, and then treat these as the true values. (Recall that we centered the X_i values so that $E(x_i) = \hat{\mu} = \overline{X} = 0$.) For the FME model parameters we used the estimates obtained from fitting the SME model since the parameter estimators we developed in (2.3) agree for the FME and SME models. While we have no explicit proof, we expect the ML parameter estimators used in this illustration would converge for $m \to \infty$ to the same quantities for both the FME and SME models.

Figure 4.2 panels (a)-(d) are analogous to Figure 4.1, but assume for the first order approximations that the FME model is true. Panel (a) plots the ratios of the SME and FME predictor MSEs. Although our assumption that the FME model is true makes the FME predictor "optimal", for many counties it actually performs worse than the SME predictor with respect to the MSE. This is because the FME predictor's optimality is in the class of unbiased predictors, and both the SME and the naïve predictors are biased, so there is no mathematical contradiction. The difference in MSEs can be up to about 50% in either direction. However, there are relatively few points for which the SME MSE is more than 20% higher than the FME MSE, while there are a substantial number where the SME MSE is more than 20% lower than the FME MSE. Computing the bias and variance terms of the MSE separately reveals that for this application when the SME predictor performs worse than the FME predictor in panel (a) it is due to the bias of the former.

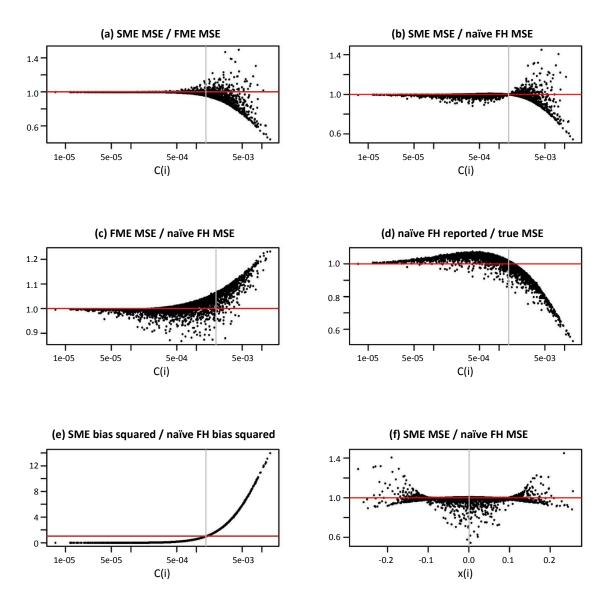


Figure 4.2 First order approximations of MSE and bias squared ratios for the U.S. county school-age children in poverty model when the FME model is true, plotted against C_i or x_i . Panels (a)-(c) show the ratios of the true MSEs of the SME and FME, SME and naïve, and FME and naïve models, respectively. Panel (d) shows the ratios of the reported MSEs and the true MSEs of the naïve model. Panel (e) shows the ratios of the biases squared of the SME and naïve models. All panels plot the ratios against C_i except panel (f), which plots the ratio of the true MSEs of the SME and naïve models against x_i . The vertical lines mark \overline{C} or \overline{x} , as appropriate.

Panel (b) of Figure 4.2 plots ratios of the MSEs of the SME and naïve predictors against C_i . It shows that when the FME model is true, the SME predictor sometimes performs better and sometimes performs worse than the naïve predictor in terms of MSE. The same statement can be made about the functional and naïve predictors based on panel (c), which shows the ratios of the FME and naïve MSEs plotted against C_i . However, as C_i increases beyond \overline{C} the FME MSE is almost always higher than the naïve predictor's MSE.

Panel (d), which plots the ratio of reported to true MSE of the naïve predictor, reminds us that the naïve model will misstate the mean squared error, sometimes overstating it and sometimes understating it. The overstatement is small, up to about 10%, but the understatement is more considerable, up to and beyond 40%. Overstatement is most likely for $C_i < \overline{C}$ and understatement for $C_i > \overline{C}$, though these tendencies do not hold for every county (as they do for the SME model) due to the variations in the squared bias terms under the FME model.

Since both the SME and naïve predictors are biased under the FME model, we can analyze the relationship between their respective biases. Panel (e) of Figure 4.2 shows the ratio of the bias squared of the SME predictors and the naïve predictors. It shows that the SME predictor has lower squared bias for $C_i < \overline{C}$, and higher squared bias for $C_i > \overline{C}$, with equality when $C_i = \overline{C}$, where the two predictors are equal. This suggests that the extreme points in the top right quadrant in panel (b) are due to the bias. The specific realization of x_i in our generation of the data will also influence these extreme points. Panel (f) plots the ratio of the SME and naïve true MSEs plotted against x_i . The vertical line represents the mean of x_i , which is approximately 0 due to how the x_i 's were generated. Note that the extreme points in the top quadrants have high deviations of x_i from its mean. On the other hand, the most extreme points in the bottom quadrants correspond to values where x_i is close to \overline{x} . This suggests large deviations of x_i 's from \overline{x} will have more impact on the true MSEs of the SME predictors than on those of the naïve predictors. For the majority of points, however, the SME model's MSEs are lower than those of the naïve model based on our first order approximations.

5 Conclusions

This paper considered three models proposed for small area estimation when one or more regression covariates are measured with error: the functional and structural measurement error models (FME and SME), and the naïve Fay-Herriot model. Section 2 established certain theoretical results for these models about parameter estimation, their small area predictions, and their corresponding prediction biases, error variances, and MSEs. This led to several observations relating the models including (i) the naïve and SME model predictions and MSEs agree, at least asymptotically, for areas with $C_i = \overline{C}$, (ii) SME prediction results converge to FME prediction results as $\sigma_x^2 \to \infty$, and (iii) in the presence of measurement error the naïve model is misspecified, so it will misstate the prediction MSE except for areas with $C_i = \overline{C}$.

Section 3 made prediction MSE comparisons between the three models over ranges of the true model's parameter values for the case when the true model was the SME. Section 4 made such comparisons taking as truth a particular SME model obtained by fitting it to data on poverty rates of school-age children for U.S. counties. This model is very similar to models used by the Census Bureau's SAIPE program, so its use provided results for a realistic case of a true SME model. MSE comparisons were also obtained for an analogous FME model by simulating values of the unobserved true covariate values x_i .

The MSE comparisons of Sections 3 and 4 tended to favor the SME model overall. Comparisons to the naïve model showed that the naïve predictor can fare poorly for C_i not near \overline{C} , with substantial MSE

increases compared to the SME model predictor. Regarding the naïve model's additional problem of misstatement of MSE for C_i not near \overline{C} , understatement of MSE when $C_i > \overline{C}$ appeared more serious than overstatement of MSE for $C_i < \overline{C}$.

From the comparisons of the SME and FME models, it was noted that when the SME model is true the FME predictor can have substantially higher prediction MSE when sampling and measurement error are large (D_i and C_i are large). While the FME predictor can be best when the FME model is true, it was also not unusual in this case for the SME and naïve FH predictors to actually have lower MSEs than the "optimal" FME predictor. This is because the optimality of the FME predictor for the FME model is among the class of unbiased predictors given fixed x_i , while the SME and naïve FH predictors, being biased, fall outside this class and so can and sometimes do have lower MSE. Though more research is needed on this point, it appears that while the avoidance of modeling assumptions for the x_i gives the FME model some potential for robustness, this can come at a significant cost in terms of higher prediction error variances for some areas.

A practical consideration related to this last point is that, in small area estimation, the most likely candidates for useful covariates with quantified measurement error (the C_i being known, or actually estimated) are other survey estimates X_i of population quantities x_i thought to be related to the population quantities θ_i whose direct estimates Y_i we seek to improve with our model. This leads to the question of why, if we believe we can adequately model θ_i through Y_i , we would choose the FME model over the SME model to avoid modeling x_i through X_i ?

Acknowledgements

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Appendix

Re-expression of MSE formulas for doing contour plots

We illustrate by showing how we re-express $\text{MSE}_s / \sigma_u^2$ in terms of $r_D = D_i / \sigma_u^2$, $r_C = C_i / \sigma_x^2$, and ρ . Given the result from Table 2.2 that $\text{MSE}_s = D_i - D_i^2 (\sigma_x^2 + C_i) / F_i^*$, we start by re-expressing F_i^* :

$$F_{i}^{*} = (\sigma_{u}^{2} + D_{i})(\sigma_{x}^{2} + C_{i}) + \beta^{2}\sigma_{x}^{2}C_{i}$$
$$= \sigma_{u}^{2}\sigma_{x}^{2}\left\{ (1 + r_{D})(1 + r_{C}) + \frac{\beta^{2}\sigma_{x}^{2}}{\sigma_{u}^{2}}r_{C} \right\}$$

From (1.7), noting that for our simplified model $\sigma_v^2 = \sigma_x^2$, we have

$$\rho^{2} = \operatorname{corr}(\theta_{i}, x_{i})^{2} = \frac{\beta^{2} \sigma_{x}^{4}}{(\sigma_{u}^{2} + \beta^{2} \sigma_{x}^{2}) \sigma_{x}^{2}} = \frac{\beta^{2} \sigma_{x}^{2}}{\sigma_{u}^{2} + \beta^{2} \sigma_{x}^{2}}$$
$$\Rightarrow r_{\rho} := \frac{\rho^{2}}{1 - \rho^{2}} = \frac{\beta^{2} \sigma_{x}^{2}}{\sigma_{u}^{2}}$$

which implies that $F_i^* = \sigma_u^2 \sigma_x^2 \left[(1 + r_D) (1 + r_C) + r_\rho r_C \right]$. Then

$$MSE_{s} = \sigma_{u}^{2} \left\{ \frac{D_{i}}{\sigma_{u}^{2}} - \frac{(1/\sigma_{u}^{2})D_{i}^{2}(\sigma_{x}^{2} + C_{i})}{\sigma_{u}^{2}\sigma_{x}^{2}\left[(1 + r_{D})(1 + r_{C}) + r_{\rho}r_{C}\right]} \right\}$$
$$\Rightarrow \frac{MSE_{s}}{\sigma_{u}^{2}} = r_{D} - \frac{r_{D}^{2}(1 + r_{C})}{(1 + r_{D})(1 + r_{C}) + r_{\rho}r_{C}}.$$

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Small area quantile estimation via spline regression and empirical likelihood

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Abstract

This paper studies small area quantile estimation under a unit level non-parametric nested-error regression model. We assume the small area specific error distributions satisfy a semi-parametric density ratio model. We fit the non-parametric model via the penalized spline regression method of Opsomer, Claeskens, Ranalli, Kauermann and Breidt (2008). Empirical likelihood is then applied to estimate the parameters in the density ratio model based on the residuals. This leads to natural area-specific estimates of error distributions. A kernel method is then applied to obtain smoothed error distribution estimates. These estimates are then used for quantile estimation in two situations: one is where we only have knowledge of covariate power means at the population level, the other is where we have covariate values of all sample units in the population. Simulation experiments indicate that the proposed methods for small area quantiles estimation work well for quantiles around the median in the first situation, and for a broad range of the quantiles in the second situation. A bootstrap mean square error estimator of the proposed estimators is also investigated. An empirical example based on Canadian income data is included.

Key Words: Small area quantile; Penalized spline; Empirical likelihood; Density ratio model; Nested-error regression model.

1 Introduction

Sample surveys are widely used to obtain information about totals, means, medians and other quantities of finite populations. Likewise, similar information on sub-populations such as individuals in specific areas and socio-demographic groups are also of interest. Often, a survey is designed to collect information of interest at the population level but leads to insufficient direct information on sub-populations. Because of this, estimating sub-population parameters with satisfactory precision and evaluating their accuracy pose serious challenges to statisticians. Statisticians must resort to suitable models to pool the information across small areas in order to properly estimate parameters for small areas when only small samples or no samples in these areas are available from the sample survey.

Research on small area estimation has received increased attention from both public and private sectors. As historical remarks, we refer to Fay and Herriot (1979), Battese, Harter and Fuller (1988), Prasad and Rao (1990), and Lahiri and Rao (1995) among many others. For a general review of the developments in small area estimation, we refer to Pfeffermann (2002) and Pfeffermann (2013) and the books of Rao (2003) and Rao and Molina (2015). See also Jiang and Lahiri (2006a), Jiang and Lahiri (2006b) and Jiang (2010) for recent publications.

Compared to quantiles, there are relatively more research activities on estimating small area means. Studies on small area quantile estimation are gaining ground. The M-quantile approach of Chambers and Tzavidis (2006) has achieved substantial success. This approach uses the M-quantile approach to

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characterize the conditional distributions of the response variable y given covariates \mathbf{x} . This information is then used to predict unobserved response values based on which the small area population distributions are estimated. Small area quantile estimation is a natural and welcome side-benefit. See Tzavidis and Chambers (2005), Pratesi, Ranalli and Salvati (2008), Tzavidis, Salvati and Pratesi (2008), and Salvati, Tzavidis and Pratesi (2012) for these developments.

Another approach for small area quantile estimation is proposed by Molina (2010). Let *s* and *r* be the sets of sampled and non-sampled units in a survey and y_s and y_r be vectors of corresponding response values. Under a parametric assumption on the joint distribution of y_s and y_r (or the transformed responses) they proposed to work out the conditional distribution of y_r given y_s (and other information). After having the joint distribution and therefore the conditional distribution properly estimated, they suggested sampling from the estimated conditional distribution to create an artificial but complete population with unobserved y_r filled up. The population distribution is estimated based on the completed population. This approach works well for estimating small area means and quantiles. Other methods we are aware of include Tzavidis, Marchetti and Chambers (2010), Chaudhuri and Ghosh (2011) and Chen and Liu (2018). Tzavidis et al. (2010) proposed a general framework for robust small area estimation, based on representing a small area estimator as a function of a predictor of this small area cumulative distribution function. Chaudhuri and Ghosh (2011) proposed an empirical likelihood based Bayesian method. Chen and Liu (2018) proposed an approach for populations admitting a nested-error linear regression model combined with error distributions satisfying a semi-parametric density ratio model (DRM). Simulations indicate that the DRM-based method stands out when the error distributions are skewed.

In this paper, we are interested in the situation where the regression function is not linear, although the nested-error regression model remains appropriate similar to Opsomer et al. (2008). Clearly, methods derived under linear models may lead to substantial bias if the linearity assumption is violated. To reduce the potential risk of serious bias, Opsomer et al. (2008) proposed an Empirical Best Linear Unbiased Prediction (EBLUP) for the small area means under a non-parametric regression model via penalized splines (P-splines); Jiang, Ngueyen and Rao (2010) developed an adaptive fence approach employing a non-parametric model selection technique; Sperlich and José Lombardía (2010) used the local polynomial inference method in the context of small area estimation; Rao, Sinha and Dumitrescu (2014) proposed a unified analysis of both discrete and continuous responses under P-spline regression models.

We follow their lead and extend their results to allow non-normal error distributions in the nested-error non-parametric regression model. More specifically, we assume the nested-error non-parametric regression model but relax the small area error distribution assumption from normal to a flexible semi-parametric DRM. We use the P-splines regression approach of Opsomer et al. (2008) to fit the nonlinear regression. Empirical likelihood is then applied to estimate the parameters in the DRM based on the residuals. This leads to natural area specific error distribution estimation. A kernel method is then applied to obtain

smoothed estimates of error distributions and small area quantiles. We construct quantile estimates in two situations: one is where we have knowledge of only covariate power means at the population level, the other is where we have covariate values of all sample units in the population. Our approach should inherit the merits of working under a non-parametric regression model, and gain from avoiding a parametric error distribution assumption. The resulting small area quantile estimates are hence more robust. Simulations indicate that when the regression function is approximately linear, the performance of the proposed approach is competitive. The proposed approach outperforms when the regression relationship is quadratic or exponential.

The rest of the paper is organized as follows. Section 2 introduces the model and assumptions. Section 3 presents the proposed approach. Section 4 proposes a bootstrap procedure for estimating mean squared errors. In Section 5, we use Monte Carlo methods to evaluate the performance of the proposed method and compare it with some existing methods. An application example is reported in Section 6. Section 7 contains some concluding remarks.

2 Model and assumptions

Consider a finite population containing $N = \sum_{i=0}^{m} N_i$ sample units partitioned into m + 1 small areas $\{(x_{ij}, y_{ij}): j = 1, 2, ..., N_i\}, i = 0, 1, ..., m$. Consider a nested-error non-parametric regression model with one covariate:

$$y_{ij} = m_0 \left(x_{ij} \right) + v_i + \varepsilon_{ij}, \qquad (2.1)$$

where x_{ij} is an auxiliary variable, v_i denotes an area-specific random effect and ε_{ij} are random errors. The regression function $m_0(\cdot)$ is unspecified, but can be approximated sufficiently well by a spline function

$$m_0\left(x;\,\boldsymbol{\beta},\,\boldsymbol{\gamma}\right) = \boldsymbol{\beta}_0 + \boldsymbol{\beta}_1 x + \ldots + \boldsymbol{\beta}_p x^p + \sum_{k=1}^K \boldsymbol{\gamma}_k \left(x - \boldsymbol{\kappa}_k\right)_+^p. \tag{2.2}$$

Here *p* is the degree of the spline, $x_{+}^{p} = x^{p}$ when x > 0 and 0 otherwise, κ_{k} , k = 1, ..., K are a set of fixed constants called knots, $\boldsymbol{\beta} = (\beta_{0}, ..., \beta_{p})'$ is a coefficient vector of the parametric portion of the model, and $\boldsymbol{\gamma} = (\gamma_{1}, ..., \gamma_{K})'$ is the vector of spline coefficients, *K* is the number of spline knots. If knot locations cover the range of *x* and *K* is sufficiently large, the class of P-spline (2.2) can approximate any smooth function $m_{0}(\cdot)$ with a high degree of accuracy, even with a small *p* (Boor, 2001). Ruppert, Wand and Carroll (2003) recommended using the number of spline knots *K* as the minimum of 40 and the number of unique *x*'s divided by 4.

We assume that a random sample from the population is obtained under an uninformative sampling plan such that (2.1) remains valid for the sampled units. Our immediate task is to fit this model based on the sampled data and we follow the approach of Opsomer et al. (2008). For ease of presentation, we first introduce some matrix notation. Let n_i be the number of units sampled from small area *i*. The response values from the *i*th area will be denoted as $\mathbf{y}_i = (y_{i1}, y_{i2}, \dots, y_{in_i})'$. We then pile them up to form the response vector of length $n: \mathbf{Y}'_n = (\mathbf{y}'_0, \mathbf{y}'_1, \dots, \mathbf{y}'_m)$. We similarly define $\boldsymbol{\epsilon}_i$ and $\boldsymbol{\epsilon}_n$ for error term. We use $\mathbf{v} = (v_0, \dots, v_m)'$ for area specific random effects and create a matrix \mathbf{D} such that

$$\mathbf{Dv} = (v_0 \mathbf{1}'_{n_0}, v_1 \mathbf{1}'_{n_1}, \dots, v_m \mathbf{1}'_{n_m})$$

with $\mathbf{1}_k$ being a length k vector of 1's. We further construct matrices \mathbf{X}_n and \mathbf{Z}_n so that their rows are made up of

$$\mathbf{x}'_{ij} = (1, x_{ij}, \dots, x^{p}_{ij}), \quad \mathbf{z}'_{ij} = ((x_{ij} - \kappa_{1})^{p}_{+}, \dots, (x_{ij} - \kappa_{K})^{p}_{+})$$

in a proper order. With these matrices and vectors, the data in the sample under model (2.1) are connected by

$$\mathbf{Y}_{n} = \mathbf{X}_{n}\boldsymbol{\beta} + \mathbf{Z}_{n}\boldsymbol{\gamma} + \mathbf{D}\mathbf{v} + \boldsymbol{\epsilon}_{n}.$$
(2.3)

Opsomer et al. (2008) fitted this model under the assumption that the components of γ , of **v** and ϵ are all independent and identically normally distributed with variances σ_{γ}^2 , σ_{ν}^2 and σ_{ϵ}^2 respectively. The solutions to the fit are given by

$$\begin{split} \hat{\mathbf{V}} &= \mathbf{Z}_{n} \mathbf{\Sigma}_{\gamma} \mathbf{Z}'_{n} + \mathbf{D} \hat{\mathbf{\Sigma}}_{\nu} \mathbf{D}' + \hat{\mathbf{\Sigma}}_{\varepsilon}, \\ \hat{\mathbf{v}} &= \hat{\mathbf{\Sigma}}_{\nu} \mathbf{D}' \hat{\mathbf{V}}^{-1} \left(\mathbf{Y}_{n} - \mathbf{X}_{n} \hat{\boldsymbol{\beta}} \right), \\ \hat{\boldsymbol{\beta}} &= \left(\mathbf{X}'_{n} \hat{\mathbf{V}}^{-1} \mathbf{X}_{n} \right)^{-1} \left(\mathbf{X}'_{n} \hat{\mathbf{V}}^{-1} \mathbf{Y}_{n} \right), \\ \hat{\boldsymbol{\gamma}} &= \hat{\mathbf{\Sigma}}_{\gamma} \mathbf{Z}'_{n} \hat{\mathbf{V}}^{-1} \left(\mathbf{Y}_{n} - \mathbf{X}_{n} \hat{\boldsymbol{\beta}} \right) \end{split}$$

where $\hat{\Sigma}_{\gamma}$, $\hat{\Sigma}_{\nu}$, $\hat{\Sigma}_{\varepsilon}$ are restricted maximum likelihood estimates of the covariance matrices of γ , \mathbf{v} and ϵ , and $\hat{\mathbf{V}}$ is the estimate of $\mathbf{V} \equiv \operatorname{var}(\mathbf{Y}_n)$.

Opsomer et al. (2008) then gave the empirical best linear unbiased predictor of the small area mean:

$$\hat{Y}_{i} = \hat{\beta}_{0} + \hat{\beta}_{1}\overline{X}_{i} + \ldots + \hat{\beta}_{p}\overline{X}_{i}^{p} + \overline{\mathbf{z}}_{i}\hat{\boldsymbol{\gamma}} + \hat{\boldsymbol{v}}_{i}, \qquad (2.4)$$

where $\overline{X}_i, ..., \overline{X}_i^p$ are the means of the powers of population units x_{ij} in area *i*, i.e., $\overline{X}_i^s = N_i^{-1} \sum_{j=1}^{N_i} x_{ij}^s$ for s = 1, ..., p, and $\overline{z}_i \hat{\gamma}$ stands for the true means of the spline basis functions over the small area *i*. Clearly, the above discussion easily extends to non-parametric additive models with two or more covariates (Lin and Zhang (1999), Ruppert et al. (2003) and Wood (2006)).

In this paper, we follow Opsomer et al. (2008) to get all the fitted values. For small area quantile estimation, we remove the normality assumption on ϵ_{ij} . Instead, we assume that their distributions $G_i(u)$ satisfy a DRM so that for i = 1, ..., m,

$$\log\left\{ dG_{i}\left(u\right)/dG_{0}\left(u\right)\right\} = \theta_{i}^{\prime}\mathbf{q}\left(u\right),\tag{2.5}$$

with a pre-specified basis function $\mathbf{q}(u)$ and an area-specific tilting parameter θ_i . One may include i = 0 in the above equation by setting $\theta_0 = 0$. We require the first element of $\mathbf{q}(u)$ to be one, so that the first element of θ_i is a normalization parameter. The DRM includes normal, Gamma, and many other distribution families as special cases. Discussions about DRM can be found in Anderson (1979), Qin and Zhang (1997), Kezioua and Leoni-Aubina (2008) and Chen and Liu (2013).

Equations (2.1), (2.2) and (2.5) together form the platform of this paper for small area quantile estimation. Our work differs from Opsomer et al. (2008) in that we focus on small area quantile estimation without a normality assumption on G_i (·). At the same time, this paper differs from Chen and Liu (2018) by postulating a non-parametric regression relationship between y_{ii} and x_{ii} instead of a linear one.

3 Proposed approach

For any $\alpha \in (0, 1)$, the α^{th} quantile of a distribution F is defined to be

$$\xi_{\alpha} = \inf \left\{ u \colon F(u) \ge \alpha \right\}$$

If $\hat{F}(u)$ is an estimate of F(u), its α -quantile is naturally estimated by

$$\hat{\xi}_{\alpha} = \inf \left\{ u \colon \hat{F}(u) \ge \alpha \right\}. \tag{3.1}$$

Under the distributional assumption on ϵ_{ii} , we have

$$P(y_{ij} \le u) = \mathbb{E} \left\{ P(\varepsilon_{ij} \le u - m_0(x_{ij}) - v_i | x_{ij}, v_i) \right\}$$
$$= \mathbb{E} \left\{ G_i(u - m_0(x_{ij}) - v_i) \right\}.$$

Hence, the population distribution of the i^{th} small area is given by

$$F_{i}(u) = N_{i}^{-1} \sum_{j=1}^{N_{i}} G_{i}(u - m_{0}(x_{ij}) - v_{i}).$$

Once G_i and $m_0(\cdot)$ are suitably estimated, so will be the small area quantiles.

We follow the empirical likelihood idea of Chen and Liu (2018) for estimating $G_i(\cdot)$. Suppose the values of ε_{ij} in the sample are known. Consider a candidate G_0 of the form

$$G_{0}(u) = \sum_{i,j} p_{ij} I\left(\varepsilon_{ij} \leq u\right),$$

where $I(\cdot)$ is an indicator function and $\sum_{i,j} = \sum_{i=0}^{m} \sum_{j=1}^{n_i}$. We hence have $p_{ij} = dG_0(\varepsilon_{ij})$ and under DRM $dG_i(\varepsilon_{st}) = p_{st} \exp\{\Theta_i' \mathbf{q}(\varepsilon_{st})\}$ for i = 0, 1, ..., m which implies

$$G_{i}(u) = \sum_{s,t} p_{st} \exp\left\{\boldsymbol{\theta}_{i}^{\prime} \mathbf{q}\left(\varepsilon_{st}\right)\right\} I\left(\varepsilon_{st} \leq u\right).$$
(3.2)

By Owen (2001), we obtain the empirical likelihood function

$$L_{n}(G_{0}, G_{1}, ..., G_{m}) = \prod_{i,j} dG_{i}(\varepsilon_{ij}) = \left\{\prod_{i,j} p_{ij}\right\} \exp\left[\sum_{i,j} \left\{ \boldsymbol{\theta}_{i}^{\prime} \boldsymbol{q}(\varepsilon_{ij})\right\}\right],$$

where the parameter $\boldsymbol{\theta}$ and p_{ij} 's satisfy $p_{ij} \ge 0$, and for s = 0, 1, ..., m,

$$\sum_{i,j} p_{ij} \exp\left\{\boldsymbol{\theta}_{s}^{\prime} \mathbf{q}\left(\varepsilon_{ij}\right)\right\} = 1.$$
(3.3)

Note that we have used the convention $\mathbf{\theta}_0 = 0$ for simpler presentation. Because G_1, \ldots, G_m are fully determined by $\mathbf{\theta}' = (\mathbf{\theta}'_1, \ldots, \mathbf{\theta}'_m)$ and G_0 , we write the empirical log-likelihood as

$$\ell_{n}\left(\boldsymbol{\theta}, G_{0}\right) = \sum_{i, j} \log\left(p_{ij}\right) + \sum_{ij} \boldsymbol{\theta}_{i}^{\prime} \mathbf{q}\left(\varepsilon_{ij}\right).$$

Maximizing $\ell(\mathbf{0}, G_0)$ with respect to G_0 under the constraints (3.3) results in fitted probabilities

$$\hat{p}_{ij} = n^{-1} \left\{ 1 + \sum_{s=1}^{m} \lambda_s \left[\exp\left\{ \Theta'_s \mathbf{q}(\varepsilon_{ij}) \right\} - 1 \right] \right\}^{-1}$$
(3.4)

and the profile log EL

$$\ell_{n}\left(\boldsymbol{\theta}\right) = -\sum_{i,j}\log\left\{1 + \sum_{s=1}^{m}\lambda_{s}\left[\exp\left\{\boldsymbol{\theta}_{s}^{\prime}\boldsymbol{q}\left(\boldsymbol{\varepsilon}_{ij}\right)\right\} - 1\right]\right\} + \sum_{i,j}\boldsymbol{\theta}_{i}^{\prime}\boldsymbol{q}\left(\boldsymbol{\varepsilon}_{ij}\right)$$

with $(\lambda_1, \ldots, \lambda_m)$ being the solution to

$$\sum_{s,t} \frac{\exp\left\{\boldsymbol{\theta}_{i}^{\prime} \mathbf{q}\left(\varepsilon_{st}\right)\right\} - 1}{1 + \sum_{l=1}^{m} \lambda_{l} \left[\exp\left\{\boldsymbol{\theta}_{l}^{\prime} \mathbf{q}\left(\varepsilon_{st}\right)\right\} - 1\right]} = 0.$$

Since the values of ε_{ij} are not available, we replace them by the residuals obtained from fitting model (2.1) under assumption (2.2):

$$\hat{\varepsilon}_{ij} = y_{ij} - \hat{m}_0 \left(x_{ij}; \hat{\beta}, \hat{\gamma} \right) - \hat{v}_i$$

where

$$\hat{m}_{0}(x;\,\hat{\beta},\,\hat{\gamma}) = \hat{\beta}_{0} + \hat{\beta}_{1}x + \ldots + \hat{\beta}_{p}x^{p} + \sum_{k=1}^{K}\hat{\gamma}_{k}\left(x - \kappa_{k}\right)_{+}^{p}.$$
(3.5)

Let $\hat{\ell}_n(\boldsymbol{\theta})$ be the log EL function $\tilde{\ell}_n(\boldsymbol{\theta})$ after ε_{ij} are replaced by $\hat{\varepsilon}_{ij}$. We define the maximum EL estimator of $\boldsymbol{\theta}$ by $\hat{\boldsymbol{\theta}} = \operatorname{argmax} \hat{\ell}_n(\boldsymbol{\theta})$ and estimate $G_i(u)$ by

$$\tilde{G}_{i}(u) = \sum_{s,t} \hat{p}_{st} \exp\left\{\hat{\boldsymbol{\theta}}_{i}^{\prime} \mathbf{q}\left(\hat{\varepsilon}_{st}\right)\right\} I\left(\hat{\varepsilon}_{st} \leq u\right)$$
(3.6)

with

$$\hat{p}_{st} = n^{-1} \left\{ 1 + \sum_{l=1}^{m} (n_l / n) \left[\exp \left\{ \theta'_l \mathbf{q} \left(\hat{\varepsilon}_{st} \right) \right\} - 1 \right] \right\}^{-1}$$

and $\hat{\theta}_0 = 0$. The R package **drmdel** can be used to compute $\hat{\theta}$ and \hat{p}_{ij} which has 11 choices of basis function $\mathbf{q}(u)$.

Because $\tilde{G}_i(u)$ is discrete, the following kernel smoothed distribution $\hat{G}_i(u)$ leads to better quantile estimation:

$$\hat{G}_{i}\left(u\right) = \sum_{j=1}^{n_{i}} \hat{w}_{ij} \Phi\left(\frac{\hat{\varepsilon}_{ij} - u}{b}\right), \tag{3.7}$$

where the weights are chosen to be $\hat{w}_{ij} = \tilde{G}_i(\hat{\varepsilon}_{ij}) - \tilde{G}_i(\hat{\varepsilon}_{ij} -)$, *b* is a bandwidth parameter, and $\Phi(\cdot)$ is the distribution function of standard normal. As suggested by Chen and Liu (2013), we choose $b = 1.06n^{-1/5} \min{\{\hat{\sigma}, \hat{Q}/1.34\}}$ where $\hat{\sigma}$ is the standard deviation of the distribution \hat{G}_i and \hat{Q} is its interquartile range.

In some applications, only population power means of covariates are known and can be used for statistical inference. In other applications, covariates of all members of the population are known. This leads two possible quantile estimates. In the first case, we estimate F_i by

$$\hat{F}_{i}^{(a)}(u) = n_{i}^{-1} \sum_{j=1}^{n_{i}} \hat{G}_{i}\left(u - \hat{\overline{Y}}_{i} - \left\{\hat{m}_{0}\left(x_{ij};\,\hat{\beta},\,\hat{\gamma}\right) - \hat{m}_{0}\left(\overline{x}_{i};\,\hat{\beta},\,\hat{\gamma}\right)\right\}\right),\tag{3.8}$$

where we use $\hat{m}_0(\bar{x}_i; \hat{\beta}, \hat{\gamma})$ specified in (3.5).

When the census information about x is available, we estimate F_i by

$$\hat{F}_{i}^{(b)}(u) = N_{i}^{-1} \left\{ \sum_{j \in s_{i}} I\left(y_{ij} \leq u\right) + \sum_{j \in r_{i}} \hat{G}_{i}\left(u - \hat{m}_{0}\left(x_{ij}\right) - \hat{v}_{i}\right) \right\},$$
(3.9)

where s_i and r_i are sets of observed and unobserved units in small area *i*. The rest of the specifications are the same as in (3.8).

The proposed estimates resemble those of Chen and Liu (2018) but we use a non-parametric regression. Because collecting population power means of covariates is easier than collecting covariates values of all units in the population $\hat{F}_i^{(a)}(u)$ is more broadly applicable than $\hat{F}_i^{(b)}(u)$. It is also computationally more efficient. Because $\hat{F}_i^{(b)}(u)$ uses covariate values of all units in the population, it should statistically outperform when both are applicable.

4 Bootstrap estimation of the mean squared errors

The proposed small area quantile estimators are assembled with many intermediate steps. It is difficult to analytically evaluate the variances or mean squared error (MSE) of such estimators. We follow others

(Sinha and Rao (2009), Tzavidis et al. (2010) and Chen and Liu (2018)) to develop a bootstrap procedure as follows:

Step 1 Obtain estimates $\hat{\boldsymbol{\beta}}$, $\hat{\boldsymbol{\gamma}}$, $\hat{\sigma}_{v}^{2}$ and $\hat{m}_{0}\left(x, \hat{\boldsymbol{\beta}}, \hat{\boldsymbol{\gamma}}\right)$ based on Model (2.1), and calculate $\hat{G}_{i}\left(u\right)$ as in (3.7).

Step 2 Generate a bootstrap finite population $H^* = \{y_{ij}^*, x_{ij}\}, i = 0, ..., m, j = 1, ..., N_i$ with $y_{ij}^* = \hat{m}_0 (x_{ij}, \hat{\beta}, \hat{\gamma}) + v_i^* + \varepsilon_{ij}^*,$

where the bootstrap residuals ε_{ij}^* are sampled from CDF $\hat{G}_i(u)$, and v_i^* are generated from $N(0, \hat{\sigma}_v^2)$.

- Step 3 From the bootstrap population H^* , we select $n_i^* = n_i$ sample units from small area *i* by simple random sampling without replacement, and repeat it *L* times to get h_i^* , l = 1, ..., L. For each sample h_i^* , compute the estimates $\hat{F}_i^{(a)^{*l}}(u)$ and $\hat{F}_i^{(b)^{*l}}(u)$ as in (3.8) and (3.9) respectively.
- Step 4 Compute the empirical MSE estimator of $\hat{\tau}$ as

mse
$$(\tau^*) = L^{-1} \sum_{l=1}^{L} (\hat{\tau}^{*l} - \tau^*)^2$$
,

where $\hat{\tau}^{*_l} = \tau(\hat{F}^{*_l}(u))$ denotes any functional of $\hat{F}^{(a)*_l}(u)$ or $\hat{F}^{(b)*_l}$ and $\tau^* = \tau(F^*(u))$ with $F^*(u)$ being the known CDF of the bootstrap populations.

Step 5 Repeat Steps 2 to 4, B times, and define the bootstrap MSE estimate as

$$B^{-1}\sum_{b=1}^{b} \operatorname{mse}(\tau^*)_b,$$

where mse $(\tau^*)_b$ is the mse (τ^*) calculated in the b^{th} repetition.

The performance of the bootstrap MSE estimator will be examined and reported in the simulation section.

5 Monte Carlo simulations

In this section, we use simulation to evaluate the performances of the proposed penalized spline regression model based empirical likelihood estimators (PEL) and their MSE estimates. When only the covariate population means are known the proposed estimators are compared with only the nested-error linear regression model based empirical likelihood estimator (LEL) of Chen and Liu (2018), and the direct estimator (DE). When covariate values are known for all sample units, the comparison is extended to also include six estimators of Tzavidis et al. (2010), denoted as EBLUP/naïve, EBLUP/CD, EBLUP/RKM, M-quantile/naïve, M-quantile/CD and M-quantile/RKM. Here, EBLUP/CD and M-quantile/CD denote the EBLUP and M-quantile estimator are obtained based on the CDF proposed by Chambers and Dunstan

(1986), and corresponding estimators based on the CDF proposed by Rao, Kovar and Mantel (1990) write as RKM.

Similar to Chen and Liu (2018), we must choose $\mathbf{q}(u)$ in the DRM. Two candidates are $\mathbf{q}_1(u) = (1, u)'$ and $\mathbf{q}_2(u) = (1, \operatorname{sign}(u)\sqrt{|u|})'$. Some preliminary simulation results indicate that $\mathbf{q}_1(u) = (1, u)'$ works well for the P-splines fitted non-parametric regression model, but $\mathbf{q}_2(u)$ does not. Instead, the choice of $\mathbf{q}_2^*(u) = (1, u, u^2)'$ leads to competitive performance. So, we use $\mathbf{q}_1(u)$ and $\mathbf{q}_2^*(u)$ in our simulation.

Following Rao et al. (2014) and Torabi and Shokoohi (2015), we generated data from the following three models:

A:
$$y_{ij} = 1 + x_{ij} + v_i + \varepsilon_{ij}$$
,
B: $y_{ij} = 1 + x_{ij} + x_{ij}^2 + v_i + \varepsilon_{ij}$,
C: $y_{ij} = 1 - x_{ij} + 0.5 \exp(x_{ij}) + v_i + \varepsilon_{ij}$.

They lead to linear, quadratic and exponential regression functions respectively. We set the number of small areas to be 30 and area population sizes $N_i = 500 (i + 1)$, i = 0, 1, ..., 29. We generated covariate x_{ij} from N(0, 1). Once x_{ij} are generated, we treated them as fixed in the simulation. The area-specific random effect v_i were generated from N(0, 1), and the errors ε_{ij} were generated from the following four distributions.

(i):
$$N(0, 1)$$
,
(ii): $t(3)$,
(iii): normal mixture $0.5N(-1, 1) + 0.5N(1, 1)$,
(iv): $N(0, \sigma_i^2)$, with $\sigma_i \sim U(0.5, 2)$, $i = 0, ..., 29$.

Distribution (ii) has a heavy tail, distributions (ii) and (iii) are symmetric, and distribution (iv) is heteroscedastic.

We used R = 1,000 repetitions in the simulation and drew random samples of size n = 500 without replacement from the population in each repetition. To avoid the possibility that some small areas have too few sample units, we drew n - 60 units at the population level and allocated an additional 2 units in each small area. We used R package **mgcv** for the REML method with default options for values of p and Kwhen fitting the P-spline function (2.4). We calculated estimates of the 5%, 25%, 50%, 75%, and 95% small area quantiles denoted as DE, LEL1, LEL2, PEL1, PEL2, for direct estimator, estimators of Chen and Liu (2018) and the proposed estimators using \mathbf{q}_1 (·) and \mathbf{q}_2 (·). We report their average mean squared error (AMSE) and absolute biases (ABIAS) defined below:

AMSE = {
$$R(m+1)$$
}⁻¹ $\sum_{i=0}^{m} \sum_{r=1}^{R} (\hat{\xi}_{i}^{(r)} - \xi_{i}^{(r)})^{2}$,
ABIAS = $(m+1)^{-1} \sum_{i=0}^{m} \left| R^{-1} \sum_{r=1}^{R} \hat{\xi}_{i}^{(r)} - R^{-1} \sum_{r=1}^{R} \xi_{i}^{(r)} \right|$

where $\hat{\xi}_i^{(r)}$ is either one of the quantile estimates of for the *i*th small area in the *r*th repetition. The results under Models A, B, and C are given in Tables 5.1-5.3 respectively. Both PEL and LEL are based on $\hat{F}_i^{(a)}$ and its mirror version in Chen and Liu (2018).

Under Model A, the linear model is valid. Hence, we expect LEL to be superior. According to Table 5.1, two methods are similar for the 25%, 50% and 75% quantiles. LELs outperform PELs for the 5% quantile while the comparison reverses for the 95% quantile. Both PEL and LEL outperform DE for the 25%, 50% and 75% quantiles with big margins. An overall impression is that the proposed methods still work satisfactorily.

Under Model B, the linear model breaks down mildly. Results in Table 5.2 show that the PEL estimators have lower AMSE for lower quantiles. The LELs still have low AMSE in spite of have higher ABIAS. The advantage of the proposed PEL under the non-parametric nested-error regression models focus for quantiles in middle levels. With fewer observations near extreme quantiles, the non-parametric model is hard to fit.

The linearity is seriously violated under Model C. LEL is expected to have poor performance and this is evident as shown in Table 5.3. At the same time, PELs work well for the 25%, 50% and 75% quantiles. The choice of $\mathbf{q}_2^*(u)$ also helps in general. For extreme quantiles, PELs remain unworth the trouble compared with DE.

				AMSE					ABIAS		
	α	DE	LEL1	LEL2	PEL1	PEL2	DE	LEL1	LEL2	PEL1	PEL2
Error distribution (i)	5%	0.470	0.120	0.142	0.121	0.162	0.346	0.022	0.028	0.024	0.032
	25%	0.219	0.074	0.080	0.074	0.082	0.081	0.006	0.006	0.006	0.006
	50%	0.187	0.067	0.067	0.067	0.068	0.011	0.005	0.005	0.006	0.006
	75%	0.218	0.074	0.079	0.074	0.082	0.081	0.007	0.005	0.008	0.006
	95%	0.470	0.121	0.142	0.123	0.165	0.340	0.024	0.031	0.023	0.033
Error distribution (ii)	5%	1.287	0.249	0.786	0.276	1.726	0.352	0.011	0.023	0.011	0.089
	25%	0.297	0.196	0.217	0.178	0.186	0.084	0.022	0.036	0.021	0.031
	50%	0.238	0.187	0.182	0.167	0.154	0.011	0.010	0.010	0.010	0.009
	75%	0.304	0.197	0.233	0.179	0.189	0.081	0.023	0.038	0.023	0.032
	95%	1.344	0.249	1.919	0.319	2.297	0.349	0.013	0.034	0.015	0.100
Error distribution (iii)	5%	0.636	0.165	0.199	0.163	0.234	0.408	0.008	0.013	0.008	0.019
	25%	0.340	0.132	0.147	0.133	0.152	0.109	0.010	0.007	0.011	0.008
	50%	0.306	0.128	0.128	0.130	0.132	0.014	0.007	0.007	0.007	0.007
	75%	0.340	0.133	0.151	0.134	0.156	0.108	0.011	0.009	0.012	0.008
	95%	0.651	0.168	0.205	0.166	0.243	0.410	0.010	0.016	0.010	0.022
Error distribution (iv)	5%	1.225	2.589	0.787	2.679	0.651	0.504	0.220	0.028	0.222	0.071
	25%	0.574	0.681	0.380	0.652	0.349	0.114	0.174	0.047	0.157	0.017
	50%	0.488	0.273	0.277	0.241	0.291	0.017	0.010	0.010	0.009	0.010
	75%	0.571	0.700	0.383	0.670	0.349	0.121	0.183	0.057	0.166	0.012
	95%	1.251	2.611	0.795	2.709	0.655	0.519	0.207	0.037	0.210	0.082

 Table 5.1

 AMSE and ABIAS of small area quantile estimators under Model A

				AMSE					ABIAS		
	α	DE	LEL1	LEL2	PEL1	PEL2	DE	LEL1	LEL2	PEL1	PEL2
Error distribution (i)	5%	0.524	2.998	2.991	0.404	0.439	0.382	1.520	1.502	0.017	0.019
	25%	0.474	0.182	0.183	0.259	0.262	0.177	0.118	0.123	0.018	0.017
	50%	0.865	0.907	0.951	0.215	0.219	0.092	0.785	0.791	0.031	0.031
	75%	1.963	0.985	1.170	0.817	0.825	0.132	0.602	0.616	0.021	0.021
	95%	7.850	3.083	3.783	9.163	9.193	1.200	1.159	1.185	0.251	0.251
Error distribution (ii)	5%	1.227	2.768	3.065	0.492	1.691	0.352	1.430	1.423	0.067	0.143
	25%	0.562	0.280	0.268	0.331	0.327	0.189	0.087	0.087	0.027	0.024
	50%	0.976	0.924	0.957	0.287	0.281	0.098	0.728	0.733	0.046	0.046
	75%	2.119	1.023	1.231	0.817	0.854	0.129	0.557	0.572	0.034	0.034
	95%	8.392	2.989	4.864	8.405	9.180	1.250	1.140	1.147	0.112	0.119
Error distribution (iii)	5%	0.842	2.171	2.207	0.425	0.491	0.500	1.252	1.238	0.013	0.014
	25%	0.657	0.209	0.209	0.292	0.296	0.176	0.076	0.077	0.010	0.011
	50%	0.935	0.791	0.805	0.244	0.249	0.082	0.679	0.682	0.026	0.027
	75%	1.983	0.981	1.086	0.739	0.752	0.131	0.588	0.597	0.024	0.024
	95%	8.020	2.782	3.251	8.344	8.385	1.219	1.059	1.078	0.144	0.145
Error distribution (iv)	5%	1.458	3.913	3.066	2.414	0.814	0.557	1.195	1.172	0.226	0.053
	25%	0.919	0.460	0.397	0.474	0.472	0.206	0.154	0.137	0.058	0.017
	50%	1.183	0.913	0.920	0.398	0.416	0.071	0.629	0.640	0.048	0.023
	75%	2.195	1.223	1.209	1.022	0.902	0.163	0.471	0.511	0.033	0.031
	95%	8.043	2.954	3.420	7.476	7.639	1.268	0.975	1.042	0.104	0.115

Table 5.2AMSE and ABIAS of small area quantile estimators under Model B

Table 5.3
AMSE and ABIAS of small area quantile estimators under Model C

				AMSE					ABIAS		
	α	DE	LEL1	LEL2	PEL1	PEL2	DE	LEL1	LEL2	PEL1	PEL2
Error distribution (i)	5%	0.279	1.340	1.258	0.092	0.151	0.267	0.997	0.978	0.051	0.031
	25%	0.146	0.316	0.263	0.087	0.098	0.068	0.282	0.280	0.035	0.046
	50%	0.152	0.326	0.403	0.094	0.096	0.011	0.215	0.227	0.019	0.015
	75%	0.335	0.868	1.368	0.225	0.244	0.029	0.665	0.700	0.043	0.044
	95%	7.011	0.890	6.818	27.97	27.81	0.291	0.206	0.301	1.398	1.384
Error distribution (ii)	5%	1.180	1.181	1.355	0.278	1.776	0.286	0.849	0.836	0.090	0.174
	25%	0.205	0.461	0.395	0.201	0.208	0.063	0.317	0.327	0.085	0.098
	50%	0.201	0.450	0.502	0.201	0.191	0.024	0.226	0.235	0.013	0.012
	75%	0.528	0.943	1.422	0.390	0.422	0.017	0.641	0.681	0.096	0.104
	95%	7.478	0.890	6.306	23.33	25.01	0.479	0.089	0.107	1.055	1.084
Error distribution (iii)	5%	0.438	1.063	1.004	0.157	0.240	0.349	0.826	0.803	0.065	0.034
	25%	0.299	0.328	0.289	0.158	0.181	0.120	0.158	0.161	0.009	0.020
	50%	0.305	0.364	0.409	0.174	0.179	0.013	0.151	0.157	0.035	0.029
	75%	0.428	0.709	1.035	0.275	0.308	0.077	0.499	0.524	0.015	0.017
	95%	6.718	0.974	4.704	24.79	25.04	0.232	0.321	0.378	1.336	1.325
Error distribution (iv)	5%	1.078	4.146	2.303	3.378	0.685	0.444	0.918	0.803	0.409	0.035
	25%	0.530	0.829	0.531	0.668	0.380	0.107	0.105	0.156	0.147	0.071
	50%	0.490	0.526	0.565	0.297	0.344	0.021	0.177	0.188	0.054	0.017
	75%	0.718	1.454	1.412	1.149	0.542	0.076	0.438	0.542	0.061	0.048
	95%	6.430	2.492	4.002	22.54	21.92	0.462	0.364	0.242	1.258	1.042

Next, we study estimators applicable when covariate values are known for all sample units. The simulation includes EB0, EB1, EB2, MQ0, MQ1 and MQ2 stand for EBLUP/naïve, EBLUP/CD, EBLUP/RKM, M-quantile/naïve, M-quantile/CD and M-quantile/RKM respectively. We set relatively small population sizes $N_i = 500$ to save some computation. Table 5.4 contains the AMSE of these estimators under Models A, B and C with N(0, 1) error distribution. To save space, we do not present the corresponding bias results. The simulation results show that the proposed method has lower AMSE and ABIAS (not presented) in general. It works well even for quantiles at rather extreme levels.

To save space, we pool the AMSE results for all 5 levels of quantiles in Table 5.5. The entry corresponding to A_i is the average AMSE for estimating quantiles at levels 5%, 25%, 50%, 75%, and 95% when data are generated from Model A with error distribution (i). We notice that with more detailed information on covariates, the LEL and PEL estimators are substantially more accurate compared to results in Tables 5.1-5.3. From Model A to Model C, the regression line becomes less linear. Correspondingly, the proposed quantile estimators have greater advantages against other estimators.

Now we evaluate the bootstrap MSE estimator proposed in Section 4. Because this method involves heavy computation, we confined the simulation to the estimator based on $\hat{F}_i^{(b)}(u)$ with basis function $\mathbf{q}_1(u) = (1, u)'$ and put B = 100, L = 100. We report the average ratios of the estimated MSEs and the simulated MSEs across all the small areas. The closer the ratio to one, more accurate the bootstrap MSE estimate. From Table 5.6 we can see that the average ratios close to one in majority situations except for error distribution (iv) on extreme levels of quantiles. We conclude that the bootstrap MSE estimator is generally satisfactory.

	α	EB0	EB1	EB2	MQ0	MQ1	MQ2	LEL1	LEL2	PEL1	PEL2
Model A	5%	0.477	0.123	0.501	0.536	0.127	0.499	0.128	0.146	0.078	0.110
	25%	0.139	0.073	0.154	0.198	0.074	0.154	0.073	0.078	0.065	0.073
	50%	0.061	0.066	0.124	0.119	0.066	0.124	0.066	0.066	0.064	0.064
	75%	0.145	0.074	0.149	0.204	0.074	0.149	0.074	0.080	0.066	0.073
	95%	0.491	0.125	0.394	0.552	0.129	0.395	0.126	0.146	0.079	0.113
Model B	5%	1.270	2.500	0.928	1.682	2.575	0.946	2.965	2.949	0.079	0.110
	25%	0.351	0.152	0.239	0.262	0.149	0.239	0.193	0.193	0.069	0.069
	50%	0.834	0.723	0.285	0.631	0.722	0.284	0.899	0.944	0.071	0.073
	75%	0.314	0.634	0.532	0.257	0.644	0.530	0.986	1.160	0.082	0.084
	95%	3.710	2.095	3.690	4.209	2.059	3.685	3.235	3.900	0.154	0.156
Model C	5%	0.346	0.830	0.415	0.708	0.307	0.351	1.087	1.028	0.075	0.130
	25%	0.345	0.173	0.169	0.388	0.110	0.154	0.263	0.224	0.066	0.075
	50%	0.340	0.170	0.142	0.207	0.150	0.136	0.291	0.349	0.065	0.067
	75%	0.288	0.577	0.211	0.191	0.376	0.227	0.731	1.088	0.068	0.087
	95%	2.578	11.47	8.087	5.194	14.64	11.96	0.868	4.215	0.148	0.156

 Table 5.4

 AMSE of 10 quantile estimators when all covariance values are known with N(0, 1) error distribution

Model	EB0	EB1	EB2	MQ0	MQ1	MQ2	LEL1	LEL2	PEL1	PEL2
$A_{\rm i}$	0.263	0.092	0.264	0.322	0.094	0.264	0.093	0.103	0.070	0.087
A_{ii}	0.810	1.379	1.822	0.810	1.381	1.796	0.217	0.370	0.203	0.744
$A_{ m iii}$	0.754	0.183	0.408	0.819	0.183	0.407	0.149	0.168	0.135	0.168
$A_{ m iv}$	0.687	0.186	0.399	0.746	0.188	0.399	0.281	0.196	0.256	0.164
B_{i}	1.296	1.221	1.135	1.408	1.230	1.138	1.832	1.829	0.091	0.098
B_{ii}	1.442	1.714	2.348	1.496	1.718	2.343	1.596	1.812	0.230	0.504
B_{iii}	1.270	1.081	1.357	1.348	1.088	1.351	1.399	1.521	0.163	0.179
$B_{_{iv}}$	1.346	1.177	1.315	1.436	1.183	1.317	1.565	1.701	0.205	0.166
C_{i}	0.799	2.645	1.805	1.339	3.117	2.566	0.648	1.381	0.084	0.103
C_{ii}	1.441	3.439	3.368	2.232	3.967	3.898	0.725	1.168	0.241	0.377
$C_{_{ m iii}}$	1.141	2.516	1.898	1.834	2.937	2.572	0.595	1.133	0.153	0.186
$C_{_{ m iv}}$	1.149	2.499	1.909	1.821	2.933	2.639	0.767	1.176	0.280	0.179

 Table 5.5

 Average AMSE over 5 quantiles when all covariate values are known

Table 5.6Average ratios of bootstrap MSEs and simulated MSEs

α	$A_{\rm i}$	$A_{ m ii}$	$A_{ m iii}$	$A_{ m iv}$	B _i	B _{ii}	B _{iii}	B _{iv}	C _i	C _{ii}	C _{iii}	C_{iv}
5%	1.01	1.03	1.05	0.36	1.05	0.98	1.01	0.39	0.99	1.19	1.10	0.27
25%	1.00	0.99	1.05	0.74	1.03	0.99	0.95	1.03	1.03	0.97	0.99	0.73
50%	1.06	1.04	0.97	1.10	1.01	1.03	0.96	0.99	1.09	0.96	0.97	1.03
75%	1.01	0.99	1.06	0.76	1.10	1.01	0.98	0.90	1.06	0.96	1.03	0.52
95%	1.04	1.20	1.10	0.33	0.89	1.02	1.13	1.02	0.95	1.37	1.13	0.69

6 Empirical application

We now illustrate the proposed estimators based on the data set *Survey of Labour and Income Dynamics* (SLID) provided by Statistics Canada (2014) downloaded from University of British Columbia library data centre. The data contain 147 variables and 47,705 sample units. We are grateful to Statistics Canada for making the data set available, but we do not address the original goal of the survey here. Instead, we use it as a superpopulation to study the effectiveness of the proposed small area quantile estimator.

In this study, we singled out 9 of the 147 variables. They are ttin, gender, spouse, edu, age, yrx, tweek, jobdur and tpaid, standing respectively for: total income, gender, whether living with the spouse, the highest level of education, age, years of experience, number of weeks employed, education level, months of duration of current job and total hours paid at this job. After removing units containing missing values in these 9 variables as well as those with $ttin \le 0$, we obtained a data set containing 28,302 sample units. The covariates power means at the population level are still calculated based on all available observations. We created 28 sub-populations (namely small areas) labeled as 4(k-1) + i, k = 1, 2, ..., 7,

i = 1, 2, 3, 4 based on gender-spouse-edu combinations. Here k denotes education level and i = 1, 2, 3, 4 denote male living with the spouse, female living with spouse, male not living with spouse and female not living with spouse respectively. The education levels are given as follows.

k	Highest education level
1	No more than 10 years elementary and secondary school
2	11-13 years of elementary and secondary school (but did not graduate)
3	Graduated high school
4	Some university or non-university postsecondary with no certificate
5	Non-university postsecondary or university certificate below Bachelor's
6	Bachelor's degree

7 University certificate above Bachelor's

We regarded log(ttin) as the response variable and fitted linear and additive non-parametric regressions with respect to other 5 variables. Based on the whole data, the adjusted R-square of the non-parametric fit is 0.482 which is much larger than 0.370 obtained by fitting the linear regression. This suggests that a non-parametric mixed model is a good choice. Figure 6.1 shows the fitted curves of log(ttin) with respect to these two covariates. Also, the R-square is as high as 0.483 even if the model includes only covariates age and tpaid and a random effect. These exploratory analyses prompt us to use only these two covariates in our simulation. We carried the simulation with sample sizes n = 200; 500 and 1,000. To make sampling proportions in small areas close to their sizes, we let $n_i = a_i + 2$, i = 1, ..., 28 with a_i generated from the multinomial distribution with $p_i = N_i/N$.

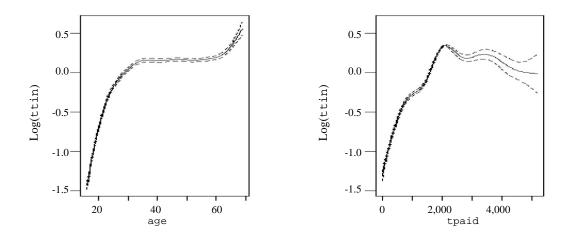


Figure 6.1 Fitted curves of log(ttin) with respect to age and tpaid.

The simulated AMSE of 10 estimators based on 1,000 repetitions are reported in Table 6.1. We first notice that both our PEL estimators outperform the other estimators, in general, indicating the advantage of our non-parametric DRM based small area estimation technique. The PEL1 compared to PEL2 has the lower AMSE for 5%, 25%, and 50% quantiles, but slightly higher AMSE for 75% and 95% quantiles indicating the heteroscedasticity of data is not serious. Regardless the PEL estimators, we notice the LEL estimators outperform other estimators for 5% quantile, and have similar performance for other quantiles. Increasing the sample size reduces the AMSE of all estimators. Clearly, it is hard to estimate the 5% quantile with a good precision because the data are skewed toward the left so there are few observations for estimating the lower quantiles. Interestingly, LEL1 is not affected as much by the skewness. We feel that the kernel smoothing step (3.7) is helpful here. Without this smoothing step, LEL1 would perform much worse. Unreported simulations show that the ABIAS of all estimators decreases in general as the sample size increases and this is most apparent for DE.

To check the performance of the proposed first estimator which using only covariate average information. In Figures 6.2, we depict the 2.5%, 50%, and 97.5% quantiles of 1,000 small area median estimates by the DE, LEL1, LEL2, PEL1, PEL2 with sample size n = 200 with the true medians marked by dots. The y-axis is the total income and x-axis is the education level. It is seen that the PEL2 boxes are the shortest for most small areas.

Table 6.2 reports the bootstrap MSE estimates as well as the average ratios of bootstrap and simulated MSEs of the small area median estimators based on $\hat{F}_i^{(a)}(u)$ and $\hat{F}_i^{(b)}(u)$ with sample size n = 200. The number of simulation repetition is 500 with basis function $\mathbf{q}_1(u) = (1, u)'$ and B = 100, L = 100. We can see the estimator $\hat{F}_i^{(a)}(u)$ has higher MSE than $\hat{F}_i^{(b)}(u)$, and most average ratios close to one.

		-									
	α	EB0	EB1	EB2	MQ0	MQ1	MQ2	LEL1	LEL2	PEL1	PEL2
<i>n</i> = 200	5%	0.784	0.769	0.901	0.714	0.763	0.885	0.245	0.421	0.242	0.336
	25%	0.107	0.256	0.488	0.102	0.261	0.467	0.115	0.131	0.097	0.152
	50%	0.080	0.119	0.236	0.064	0.116	0.223	0.076	0.095	0.056	0.102
	75%	0.122	0.100	0.142	0.085	0.102	0.138	0.085	0.076	0.069	0.068
	95%	0.233	0.190	0.280	0.141	0.138	0.266	0.217	0.179	0.117	0.096
<i>n</i> = 500	5%	0.793	0.603	0.826	0.710	0.579	0.805	0.173	0.345	0.210	0.301
	25%	0.072	0.110	0.207	0.076	0.119	0.197	0.069	0.127	0.063	0.091
	50%	0.049	0.050	0.074	0.036	0.050	0.072	0.053	0.076	0.040	0.043
	75%	0.108	0.044	0.060	0.055	0.046	0.058	0.054	0.047	0.046	0.043
	95%	0.257	0.128	0.152	0.109	0.058	0.148	0.138	0.125	0.086	0.077
<i>n</i> = 1,000	5%	0.792	0.397	0.542	0.706	0.377	0.528	0.078	0.130	0.095	0.144
	25%	0.054	0.056	0.098	0.066	0.067	0.095	0.041	0.043	0.038	0.056
	50%	0.034	0.026	0.032	0.027	0.026	0.031	0.019	0.028	0.018	0.024
	75%	0.102	0.024	0.030	0.043	0.026	0.030	0.037	0.033	0.019	0.023
	95%	0.270	0.088	0.090	0.095	0.114	0.090	0.074	0.067	0.053	0.057

 Table 6.1

 AMSE of small area quantile estimators based on real data

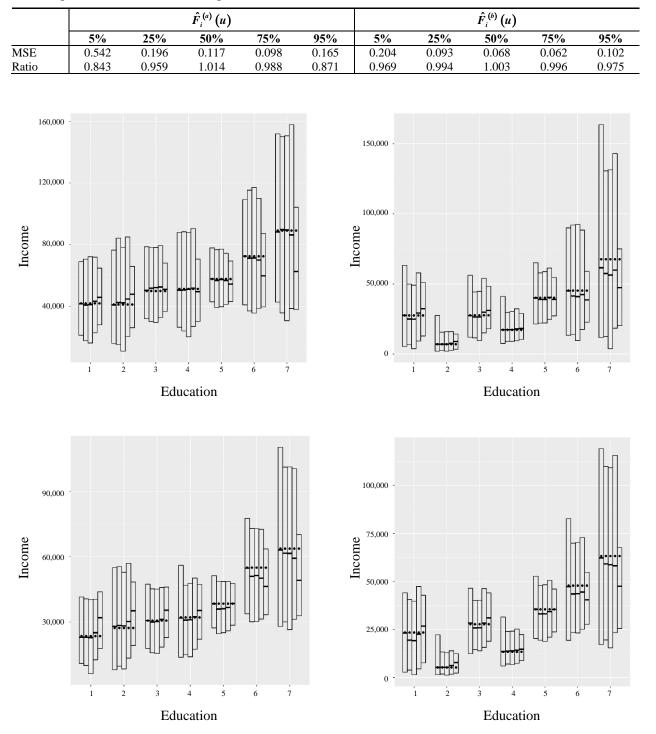


 Table 6.2

 Bootstrap MSE estimates and average ratios of the estimated and simulated MSEs

Figure 6.2 The bottom, middle and top lines of each bar denote 2.5%, 50% and 97.5% quantiles of 1,000 small area estimates of the total income. The dot in each bar denotes true small area median. Five bars in each cluster are formed by DE, LEL1, LEL2, PEL1, PEL2 estimates. Top two plots: male living (left) and not living (right) with spouse; Bottom two plots: female living (left) and not living (right) with spouse; Bottom to 7 education levels.

7 Conclusion

We studied the small area quantile estimation under the nested-error non-parametric regression model and a semi-parametric DRM assumption on error distributions. We proposed two quantile estimators based on P-splines and empirical likelihood approach. Simulation results show that the proposed estimators are robust and have respectable efficiency under both linear and non-parametric regression functions for midrange quantiles. The proposed approach can be extended to non-parametric regression models with multiple covariates in principle, though it will lead to many more parameters to be estimated. This problem will be investigated in a future work.

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Development of a small area estimation system at Statistics Canada

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Abstract

The demand for small area estimates by users of Statistics Canada's data has been steadily increasing over recent years. In this paper, we provide a summary of procedures that have been incorporated into a SAS based production system for producing official small area estimates at Statistics Canada. This system includes: procedures based on unit or area level models; the incorporation of the sampling design; the ability to smooth the design variance for each small area if an area level model is used; the ability to ensure that the small area estimates add up to reliable higher level estimates; and the development of diagnostic tools to test the adequacy of the model. The production system has been used to produce small area estimates on an experimental basis for several surveys at Statistics Canada that include: the estimation of health characteristics, the estimation of undercoverage in the census, the estimation of manufacturing sales and the estimation of unemployment rates and employment counts for the Labour Force Survey. Some of the diagnostic implemented in the system are illustrated using Labour Force Survey data along with administrative auxiliary data.

Key Words: Small area estimation; Area level model; Unit level model; EBLUP; Hierarchical Bayes methods; Official Statistics.

1 Introduction

Today's data users are becoming more and more sophisticated and are asking for more data and at more detailed levels. For National Statistical Offices (NSOs) facing declining response rates, producing data at finer levels of detail is a particularly daunting challenge. Small area estimation techniques are one way that can be considered to meet this demand to produce estimates for specified sub-populations or small areas. A *small area* refers to a subgroup of the population for which the sample size is so small that direct estimates are not reliable enough to be published. Examples of small areas include a geographical region (e.g., a province, county, municipality, etc.), a demographic group (e.g., age by sex), a demographic group within a geographic region or a detailed industry group. The demand for small area data has been recognized for years (see Brackstone, 1987), but recently, it has greatly increased as noted in the spring 2014 report of the Auditor General of Canada.

The study of small area estimation procedures has a long history at Statistics Canada, beginning in the seventies with Singh and Tessier (1976) and Ghangurde and Singh (1977). Drew, Singh and Choudhry (1982) proposed a sample dependent procedure to estimate employment characteristics below the provincial level. Dick (1995) modeled net undercoverage for the 1991 Canadian Census of Population. The development of a small area estimation system suited to Statistics Canada surveys is well-timed, as there is now a great deal of literature written on the subject, including the books by Rao (2003) and Rao and Molina (2015).

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Four papers that have had a great impact in small area estimation (SAE) are Gonzalez and Hoza (1978), Fay and Herriot (1979), Battese, Harter and Fuller (1988), and Prasad and Rao (1990). Gonzalez and Hoza (1978) were among the first to propose small area estimation procedures (mainly synthetic estimation). Fay and Herriot (1979) developed procedures to estimate income for small areas using the long form Census Data. This method and its variants are among the most widely used procedures for producing small area estimates through the integration of auxiliary data with direct survey estimates. Battese et al. (1988) developed a small area procedure to estimate crop areas using survey and satellite data available for individual units. Finally, Prasad and Rao (1990) derived a nearly unbiased estimator of the model-based mean squared error for both the Fay-Herriot and Battese-Harter-Fuller estimators.

The statistical theory of model-based SAE is rather complex and much of the software available at National Statistical Offices has been programmed on a one-time basis and, as such, is not appropriate in a production environment. It was therefore decided to develop a system as it would be beneficial as a production tool, as well as a learning tool for employees. At the time that this was decided, around 2006, there existed computer programs developed by the EURAREA (2004) project for small area estimation. However, this set of programs was no longer in development mode and did not represent the latest advances in small area estimation. Therefore, a flexible small area estimation system that would address the needs of producing small area estimates in production was developed at Statistics Canada. Some of the basic requirements of this small area system included: allowing for both area and unit level models; incorporating the sampling design in the estimation of the parameters of interest and the mean squared error; ensuring that the small area estimates would add up to reliable higher level estimates (i.e., totals), and developing diagnostic tools to test the adequacy of the models used for small area estimation. A prototype system, written in SAS, was therefore developed by Estevao, Hidiroglou and You (2015) to reflect these requirements. This prototype has been transformed into a production system that is currently used by Statistics Canada.

The paper is organized as follows. Section 2 introduces the notation used in the article. Section 3 discusses the options available in the production system for the area level model and Empirical Best Linear Unbiased Prediction (EBLUP) methods. The options for the unit level model with EBLUP methods are presented in Section 4. The Hierarchical Bayes approach is presented in Section 5 for the area level model. Section 6 illustrates the production system using Statistics Canada's Labour Force Survey. Finally, some conclusions are given in Section 7.

2 Core notation and background

We first introduce some notation that will define the various small area estimators included in the production system. Let U denote a population of size N. This population is partitioned into M mutually exclusive and exhaustive areas, where each area $U_i \subset U$, i = 1, ..., M has N_i observations. A sample, s, of size n is drawn from the population using a well-defined probability mechanism p(s) and the resulting sample is split into areas $s_i = s \cap U_i$, i = 1, ..., M. Note that, for some of the areas, the realized

sample size n_i may be zero. The set of $m(m \le M)$ areas, where n_i is strictly greater than 0, will be denoted as A. The set of the remaining areas, where n_i is equal than 0, will be denoted as \overline{A} .

Let $\pi_j = \sum_{\{s: j \in s\}} p(s)$, $j \in U$, be the inclusion probabilities where $\{s: j \in s\}$ denotes summation over all samples *s* containing unit *j*. We denote the sampling weight for unit *j* as d_j , where $d_j = \pi_j^{-1}$. The final weight associated with unit *j* will be denoted as w_j . This weight will normally be the product of the original design weight (d_j) times an adjustment factor that reflects the incorporation of available auxiliary data (via regression or calibration), as well as non-response adjustments. Note that the auxiliary data used in the adjustment factor may not necessarily be the same as those used for small area estimation.

The objective of a small area estimation system is to estimate a population parameter θ_i (e.g., a mean or a total) for each area *i* for a given variable of interest *y* when some area sample sizes n_i are too small to use *direct estimation* procedures. A *direct estimator* of θ_i is one that uses values of the variable of interest, *y*, strictly from the sample units in area *i*. However, a major disadvantage of such estimators is that unacceptably large standard errors may result: this is especially true if the area sample size is small. Small area procedures use *indirect estimators* that borrow strength across areas, by using models which link all areas through some common parameters. Indirect estimators will be efficient (i.e., increase the effective sample size and thus decrease the standard error) if the model holds for each area. Departures from the model will result in reduced accuracy. There is a wide variety of indirect estimators available and a good summary is provided in Rao and Molina (2015).

Small area estimators are classified as area or unit level depending on the level at which the modeling is performed. *Area level* small area estimators are based on models linking a given parameter of interest to area-specific auxiliary variables. *Unit level* small area estimators are based on models linking the variable of interest to unit-specific auxiliary variables. Area level small area estimators are computed if the unit level area data are not available. They can also be computed if the unit level data are available by aggregating them to the appropriate area level. This might be useful in practice because the area level small area estimators may be less prone to outliers than their unit level counterpart.

3 Area level model

The area level small area estimator first appeared in the seminal paper of Fay and Herriot (1979). Following that paper, let the parameter of interest be θ_i ; common examples are totals, $Y_i = \sum_{j \in U_i} y_j$, or means, $\overline{Y_i} = Y_i / N_i$. As noted above, the vector of auxiliary variables may differ from the one used in direct estimation and is denoted as \mathbf{z} . The area level model can be expressed as two equations.

The first equation, commonly known as the sampling model, is given by

$$\hat{\theta}_i = \theta_i + e_i \tag{3.1}$$

and expresses the direct estimate $\hat{\theta}_i$ in terms of the unknown parameter θ_i plus a random error e_i due to sampling. The sampling errors e_i are independently and identically distributed with mean 0 and variance

 ψ_i : that is $E_p(e_i | \theta_i) = 0$ and $V_p(e_i | \theta_i) = \psi_i$, where *p* denotes expectation in terms of the sample design. Note that ψ_i is also the design variance of $\hat{\theta}_i$ and is typically unknown.

The second equation, known as the linking model, is given by

$$\boldsymbol{\theta}_i = \mathbf{z}_i^T \boldsymbol{\beta} + b_i \boldsymbol{v}_i \tag{3.2}$$

and expresses the parameter θ_i as a fixed effect $\mathbf{z}_i^T \boldsymbol{\beta}$ plus a random effect v_i multiplied by b_i . In the production system, the b_i term has a default value of one but can be specified by the user to control heteroscedastic errors or the impact of influential observations. The random effects v_i are independently and identically distributed with mean 0 and unknown model variance σ_v^2 , that is $E_m(v_i) = 0$ and $V_m(v_i) = \sigma_v^2$ where E_m denotes the model expectation and V_m the model variance. The random errors e_i are independent of the random effects v_i . The combination of the sampling model and linking model results in a single generalized linear mixed model (GLMM) given by

$$\hat{\theta}_i = \mathbf{z}_i^T \mathbf{\beta} + b_i v_i + e_i. \tag{3.3}$$

From the Fay-Herriot model (3.3), we observe that $E_{mp}(\hat{\theta}_i) = \mathbf{z}_i^T \boldsymbol{\beta}$ and $V_{mp}(\hat{\theta}_i) = b_i^2 \sigma_v^2 + \tilde{\psi}_i$, where $\tilde{\psi}_i = E_m(\psi_i)$ is the smoothed design variance of $\hat{\theta}_i$. In general, we cannot treat ψ_i as fixed, as it is not strictly a function of auxiliary data. If the σ_v^2 's and $\tilde{\psi}_i$'s are known, the solution to the GLMM yields the Best Linear Unbiased Predictor (BLUP), $\tilde{\theta}_i^{\text{BLUP}}$

$$\tilde{\theta}_{i}^{\text{BLUP}} = \begin{cases} \gamma_{i}\hat{\theta}_{i} + (1 - \gamma_{i}) \mathbf{z}_{i}^{T} \tilde{\boldsymbol{\beta}} & \text{for } i \in A \\ \mathbf{z}_{i}^{T} \tilde{\boldsymbol{\beta}} & \text{for } i \in \overline{A} \end{cases}$$
(3.4)

where $\gamma_i = (b_i^2 \sigma_v^2) / (\tilde{\psi}_i + b_i^2 \sigma_v^2)$ and $\tilde{\boldsymbol{\beta}} = \left(\sum_{i \in A} \mathbf{z}_i \mathbf{z}_i^T / (\tilde{\psi}_i + b_i^2 \sigma_v^2)\right)^{-1} \sum_{i \in A} \mathbf{z}_i \hat{\theta}_i / (\tilde{\psi}_i + b_i^2 \sigma_v^2).$

There are four recursive procedures for estimating σ_v^2 and β in the production system. The first three assume that $\tilde{\psi}_i$ is known, or that a smoothed version of it is available (see the following section for details). Under this assumption, the variance components can be computed via the Fay-Herriot procedure (FH) as outlined in Fay and Herriot (1979), the restricted maximum likelihood (REML), or the Adjusted Density Maximization (ADM) due to Li and Lahiri (2010). The fourth procedure, WF, due to Wang and Fuller (2003) assumes that ψ_i is estimated by $\hat{\psi}_i$ given that $n_i \ge 2$. The WF procedure does not require any smoothing of the estimated $\hat{\psi}_i$ values before estimating σ_v^2 . Wang and Fuller (2003) carried out simulations with n_i ranging from 9 to 36 and found that their procedure yielded reasonable estimates of θ_i and its estimated mean squared error.

The main difference between these four procedures is how the σ_v^2 's are computed. They are all based on an iterative scoring algorithm that obtains $\hat{\sigma}_v^2$ as an estimate of the model variance σ_v^2 . The FH, REML, and WF procedures may yield $\hat{\sigma}_v^2$'s that are smaller than zero. If this occurs, the $\hat{\sigma}_v^2$'s are set to zero for both the FH and REML procedures. A drawback of truncating the estimated σ_v^2 to zero is that the resulting small area estimator will be synthetic for all areas. Li and Lahiri (2010) suggested the ADM as a way to address the problem of obtaining negative $\hat{\sigma}_{\nu}^2$ by maximizing an adjusted likelihood defined as a product of the model variance and a standard likelihood. Although the ADM method always gives a positive solution for σ_{ν}^2 , it should be used cautiously because it overestimates the model variance. The REML, FH and ADM procedures use the smoothed values of the estimated $\hat{\psi}_i$ values obtained from the sample or some estimate provided by the user. For the WF procedure, if $\hat{\sigma}_{\nu}^2 < 0$, Wang and Fuller (2003) suggested to set $\hat{\sigma}_{\nu}^2$ to $0.5\sqrt{\hat{V}(\hat{\sigma}_{\nu}^2)}$, where

$$\hat{V}(\hat{\sigma}_{v}^{2}) = \sum_{i \in A} 2\kappa_{i}^{2} \left[\left(\hat{\psi}_{i} + b_{i}^{2} \hat{\sigma}_{v}^{2} \right)^{2} + \frac{\left(\hat{\psi}_{i} \right)^{2}}{\left(n_{i} - 1 \right)} \right]$$

and

$$\kappa_{i} = \frac{\left[b_{i}^{2}\hat{\sigma}_{v}^{2} + \frac{(n_{i}+1)}{(n_{i}-1)}\hat{\psi}_{i}\right]^{-1}}{\sum_{i\in A} \left[b_{i}^{2}\hat{\sigma}_{v}^{2} + \frac{(n_{i}+1)}{(n_{i}-1)}\hat{\psi}_{i}\right]^{-1}}.$$

Plugging $\hat{\sigma}_{\nu}^2$ and an estimate of $\tilde{\psi}_i$'s into the $\tilde{\theta}_i^{\text{BLUP}}$, defined by equation (3.4), yields the Empirical Best Linear Unbiased Predictor (EBLUP), $\hat{\theta}_i^{\text{EBLUP}}$. It is given by

$$\hat{\theta}_{i}^{\text{EBLUP}} = \begin{cases} \hat{\gamma}_{i} \hat{\theta}_{i} + (1 - \hat{\gamma}_{i}) \mathbf{z}_{i}^{T} \hat{\boldsymbol{\beta}} & \text{for } i \in A \\ \mathbf{z}_{i}^{T} \hat{\boldsymbol{\beta}} & \text{for } i \in \overline{A} \end{cases}$$

where $\hat{\gamma}_i = (b_i^2 \hat{\sigma}_v^2) / (\ddot{\psi}_i + b_i^2 \hat{\sigma}_v^2)$, $\hat{\beta} = \left(\sum_{i \in A} \mathbf{z}_i \mathbf{z}_i^T / (\ddot{\psi}_i + b_i^2 \hat{\sigma}_v^2)\right)^{-1} \sum_{i \in A} \mathbf{z}_i \hat{\theta}_i^{\text{DIR}} / (\ddot{\psi}_i + b_i^2 \hat{\sigma}_v^2)$, and $\ddot{\psi}_i$ is chosen according to the procedure used. For the REML, FH and ADM procedures the $\ddot{\psi}_i$'s are the smoothed values of the estimated $\hat{\psi}_i$ values obtained from the sample or some estimate provided by the user. For the WF procedure, we have that $\ddot{\psi}_i = \hat{\psi}_i$. If the estimated model variance $b_i^2 \hat{\sigma}_v^2$ is relatively small compared with $\ddot{\psi}_i$, then $\hat{\gamma}_i$ will be small and more weight will be attached to the synthetic estimator $\mathbf{z}_i^T \hat{\boldsymbol{\beta}}$. Similarly, more weight is attached to the direct estimator, $\hat{\theta}_i$, if the design variance $\ddot{\psi}_i$ is relatively small.

Details of the required computations can be found in the methodology specifications for the production system in Estevao et al. (2015).

3.1 Estimation of the smooth design variance

The design variance, ψ_i , could be used as an estimator of the smooth design variance $\tilde{\psi}_i = E_m(\psi_i)$ if it were known. In most cases, it is unknown. To get around this difficulty, a design-unbiased variance estimator $\hat{\psi}_i$ of ψ_i is assumed to be available; i.e., $E_p(\hat{\psi}_i) = \psi_i$. Under this assumption, we have that

$$E_{mp}\left(\hat{\psi}_{i}\right)=E_{m}\left(\psi_{i}\right)=\tilde{\psi}_{i}.$$

A simple unbiased estimator of the smooth design variance $\tilde{\psi}_i$ is $\hat{\psi}_i$. However, $\hat{\psi}_i$ may be quite unstable when the sample size in domain *i* is small. A more efficient estimator is obtained by modelling $\hat{\psi}_i$ given \mathbf{z}_i . Dick (1995) and Rivest and Belmonte (2000) considered smoothing models given by

$$\log\left(\hat{\psi}_{i}\right) = \mathbf{x}_{i}^{T}\boldsymbol{\alpha} + \boldsymbol{\varepsilon}_{i},$$

where \mathbf{x}_i is a vector of explanatory variables that are functions of \mathbf{z}_i , $\boldsymbol{\alpha}$ is a vector of unknown model parameters to be estimated, and ε_i is a random error with $E_{mp}(\varepsilon_i) = 0$ and constant variance $\sigma_{\varepsilon}^2 = V_{mp}(\varepsilon_i)$. We also assume that the errors ε_i are identically distributed conditionally on \mathbf{z}_i , i = 1, ..., m. From the above model, we observe that

$$\tilde{\psi}_i = E_{mn}\left(\hat{\psi}_i\right) = \exp\left(\mathbf{x}_i^T\boldsymbol{\alpha}\right)\Delta,$$

where $\Delta = E_{mp} (\exp(\varepsilon_i))$. Dick (1995) estimated $\tilde{\psi}_i$ by omitting the factor Δ . Rivest and Belmonte (2000) estimated Δ by assuming that the errors ε_i are normally distributed. However, we observed empirically that the resulting estimator of Δ is sensitive to deviations from the normality assumption. This assumption is avoided by using a method of moments (see Beaumont and Bocci, 2016). This leads to the unbiased estimator of Δ given by

$$\hat{\Delta}(\boldsymbol{\alpha}) = \frac{\sum_{i=1}^{m} \hat{\psi}_i}{\sum_{i=1}^{m} \exp(\mathbf{x}_i^T \boldsymbol{\alpha})}.$$

An estimator $\hat{\alpha}$ of the vector of unknown model parameters α is necessary to estimate $\tilde{\psi}_i$. It is obtained using the ordinary least squares method as

$$\hat{\boldsymbol{\alpha}} = \left(\sum_{i=1}^{m} \mathbf{x}_{i} \mathbf{x}_{i}^{T}\right)^{-1} \sum_{i=1}^{m} \mathbf{x}_{i} \log\left(\hat{\boldsymbol{\psi}}_{i}\right).$$

The estimator $\hat{\psi}_i$ of $\hat{\psi}_i$ is then given by

$$\hat{\tilde{\psi}}_i = \exp\left(\mathbf{x}_i^T \hat{\boldsymbol{\alpha}}\right) \hat{\Delta}(\hat{\boldsymbol{\alpha}}).$$

A nice property of $\hat{\psi}_i$ is that the average of the smooth design variance estimator, $\hat{\psi}_i$, is equal to the average of the direct variance estimator, $\hat{\psi}_i$; i.e.,

$$\frac{\sum_{i=1}^{m}\hat{\psi}_{i}}{m}=\frac{\sum_{i=1}^{m}\hat{\psi}_{i}}{m}$$

This ensures that $\hat{\psi}_i$ does not systematically overestimate or underestimate $\hat{\psi}_i = E_{mp}(\hat{\psi}_i)$.

3.2 Benchmarking

If the parameter of interest θ_i is a total $(\theta_i = Y_i)$, the user may wish to have the sum of the small area estimates, $\hat{\theta} = \sum_{i \in A \cup \overline{A}} \hat{\theta}_i^{\text{EBLUP}}$, agree with the estimated totals $\hat{Y} = \sum_{i \in A} \hat{Y}_i$ at the overall sample level *s*;

i.e., $\hat{\theta} = \hat{Y}$. In the case of a mean, $\theta_i = \overline{Y}_i$, this benchmarking condition becomes $\sum_{i \in A \cup \overline{A}} N_i \hat{\theta}_i^{\text{EBLUP}} = \sum_{i \in A} N_i \hat{\theta}_i$, where $\hat{\theta}_i = \hat{Y}_i$.

Two methods are available in the production system to ensure benchmarking for area based small area estimates. The first one is based on a difference adjustment and the second one is based on an augmented vector. They are valid for any method used to compute $\hat{\theta}_i^{\text{EBLUP}}$ or whether the variance estimate $\ddot{\psi}_i$ has been smoothed or not. The benchmarking based on a difference adjustment is an adaptation of the benchmarking given in Battese et al. (1988). The benchmarking based on an augmented vector is due to Wang, Fuller and Qu (2008).

Difference adjustment: For this method, the $\hat{\theta}_i^{\text{EBLUP}}$ estimator is adjusted only for those areas where the realized sample size $n_i \ge 1$, $i \in A$ and the synthetic estimates $\mathbf{z}_i^T \hat{\boldsymbol{\beta}}$ for $i \in \overline{A}$ are left as is. The resulting benchmarked estimator is given by $\hat{\theta}_i^{\text{EBLUP}, b}$ and is defined as follows

$$\hat{\theta}_{i}^{\text{EBLUP, b}} = \begin{cases} \hat{\theta}_{i}^{\text{EBLUP}} + \alpha_{i} \left(\hat{\theta}^{*} - \sum_{d \in A} \omega_{d} \hat{\theta}_{d}^{\text{EBLUP}} \right) & \text{for } i \in A \\ \mathbf{z}_{i}^{T} \hat{\boldsymbol{\beta}} & \text{for } i \in \overline{A} \end{cases}$$

where $\alpha_i = \left\{\sum_{i \in U_A} \omega_i^2 \left(\ddot{\psi}_i + b_i^2 \hat{\sigma}_v^2\right)\right\}^{-1} \omega_i \left(\ddot{\psi}_i + b_i^2 \hat{\sigma}_v^2\right)$ for $i \in A$, $\omega_i = 1$, if the benchmarking is to a total, and $\omega_i = N_i / N$, if the benchmarking is for the mean. The estimator $\hat{\theta}^*$ is a value provided by the user that represents the total or mean of the *y*-values of population *U*. The benchmarking ensures that $\sum_{i \in A \cup \overline{A}} \omega_i \hat{\theta}_i^{\text{EBLUP, b}} = \hat{\theta}^*$.

Augmented vector: The vector \mathbf{z}_i^T is augmented with $\omega_i \dot{\psi}_i$, to form $\mathbf{z}_i^{*T} = (\mathbf{z}_i^T, \omega_i \dot{\psi}_i)$ with ω_i and $\ddot{\psi}_i$ as previously defined. The resulting augmented generalized linear mixed model (GLMM) equation is given by

$$\hat{\theta}_i = \mathbf{z}_i^{*T} \mathbf{\beta}^* + b_i v_i^* + e_i \tag{3.5}$$

where $E_m(v_i^*) = 0$ and $V_m(v_i^*) = \sigma_v^{*2}$. The estimates for β^* and σ_v^{*2} are once more solved recursively for the four EBLUP procedures that we denote as $\hat{\theta}_i^{\text{EBLUP*}}$.

The resulting benchmarked estimator $\hat{\theta}_i^{\text{EBLUP}^*, b}$ is given by

$$\hat{\theta}_{i}^{\text{EBLUP}^{*}, b} = \begin{cases} \hat{\gamma}_{i}^{*} \hat{\theta}_{i}^{\text{EBLUP}^{*}} + (1 - \hat{\gamma}_{i}^{*}) \mathbf{z}_{i}^{*T} \hat{\boldsymbol{\beta}}^{*} & \text{for } i \in A \\ \mathbf{z}_{i}^{*T} \hat{\boldsymbol{\beta}}^{*} & \text{for } i \in \overline{A} \end{cases}$$

where $\hat{\gamma}_{i}^{*} = (b_{i}^{2}\hat{\sigma}_{v}^{*2})/(\ddot{\psi}_{i} + b_{i}^{2}\hat{\sigma}_{v}^{*2})$, and $\hat{\beta}^{*} = (\sum_{i \in A} \mathbf{z}_{i}^{*}\mathbf{z}_{i}^{*T}/(\ddot{\psi}_{i} + b_{i}^{2}\hat{\sigma}_{v}^{*2}))^{-1} \sum_{i \in A} \mathbf{z}_{i}^{*}\hat{\theta}_{i}/(\ddot{\psi}_{i} + b_{i}^{2}\hat{\sigma}_{v}^{*2})$.

All the components of $\hat{\theta}_i^{\text{EBLUP}^*, b}$ are computed using the augmented model given by (3.5). It can be shown that $\sum_{i \in A \cup \overline{A}} \omega_i \hat{\theta}_i^{\text{EBLUP}^*, b} = \sum_{i \in A} \omega_i \hat{\theta}_i$, and hence the benchmarking holds.

The difference adjustment and augmented vector methods are two ways that benchmarking can be satisfied. Wang et al. (2008) suggested other procedures that can be used. Specifically, they adapted the self-calibrated estimator You and Rao (2002) developed in the context of the unit level model to the area

level model. You, Rao and Hidiroglou (2013) obtained an estimator of the mean squared prediction error and its bias under a misspecified model.

3.3 Mean squared error estimation

The reliability of the EBLUP estimators is obtained as $MSE(\hat{\theta}_i^{EBLUP}) = E(\hat{\theta}_i^{EBLUP} - \theta_i)^2$. The expectation is with respect to models (3.3) for the non-benchmarked estimator, and (3.5) for the benchmarked estimator.

The estimated Mean Squared Errors (MSEs) of the area level estimators are given in Table 3.1. The specific form of the *g* terms and the estimated variances can be found in Rao and Molina (2015) or in Estevao et al. (2015). For the benchmarked estimators, the estimated MSE for the difference adjustment approach uses the non-benchmarked MSE formulas. For the case of the augmented vector approach, the MSE is based on augmenting the vector \mathbf{z}_i^T with $\omega_i \ddot{\psi}_i$.

Estimator	mse
Fay-Herriot	$\operatorname{mse}\left(\hat{\theta}_{i}^{\operatorname{FH}}\right) = \begin{cases} g_{0i} + g_{1i} + g_{2i} + 2g_{3i} & \text{for } i \in A \\ \mathbf{z}_{i}^{T} \operatorname{var}\left(\hat{\boldsymbol{\beta}}\right) \mathbf{z}_{i} + b_{i}^{2} \hat{\sigma}_{v}^{2} & \text{for } i \in \overline{A} \end{cases}$
ADM	$\operatorname{mse}\left(\hat{\theta}_{i}^{\mathrm{ADM}}\right) = \begin{cases} g_{0i} + g_{1i} + g_{2i} + 2g_{3i} & \text{for } i \in A \\ \mathbf{z}_{i}^{T} \operatorname{var}\left(\hat{\boldsymbol{\beta}}\right) \mathbf{z}_{i} + b_{i}^{2} \hat{\sigma}_{v}^{2} & \text{for } i \in \overline{A} \end{cases}$
REML	$\operatorname{mse}\left(\hat{\theta}_{i}^{\operatorname{REML}}\right) = \begin{cases} g_{1i} + g_{2i} + 2g_{3i} & \text{for } i \in A \\ \mathbf{z}_{i}^{T}\operatorname{var}\left(\hat{\boldsymbol{\beta}}\right)\mathbf{z}_{i} + b_{i}^{2}\hat{\sigma}_{v}^{2} & \text{for } i \in \overline{A} \end{cases}$
WF	$\operatorname{mse}\left(\hat{\theta}_{i}^{\operatorname{WF}}\right) = \begin{cases} g_{1i} + g_{2i} + 2g_{3i} + g_{4i} & \text{for } i \in A \\ \mathbf{z}_{i}^{T} \operatorname{var}\left(\hat{\boldsymbol{\beta}}\right) \mathbf{z}_{i} + b_{i}^{2} \hat{\sigma}_{v}^{2} & \text{for } i \in \overline{A} \end{cases}$

Table 3.1MSE estimates (mse) for the area level estimators

The various *g* terms in Table 3.1 can be interpreted as follows. The g_{0i} is a bias correction term for FH and ADM. The g_{1i} term given by $g_{1i} = \hat{\gamma}_i \ddot{\psi}_i$, accounts for most of the MSE if the number of areas is large. The g_{2i} term accounts for the estimation of $\boldsymbol{\beta}$, and $2g_{3i}$ accounts for the estimation of σ_v^2 . The g_{4i} term in the WF procedure reflects that the estimated value of ψ_i , $\hat{\psi}_i$, has been used. The estimated variance of $\hat{\boldsymbol{\beta}}$, given by $\operatorname{var}(\hat{\boldsymbol{\beta}}) = \left(\sum_{i \in A} \frac{z_i z_i^T}{\psi_i + b_i^2 \hat{\sigma}_v^2}\right)^{-1}$ is dependent on the particular procedure used to estimate σ_v^2 .

4 Unit level model

The original unit level model was proposed by Battese et al. (1988). They assumed the following nested error model

$$y_{ij} = \mathbf{z}_{ij}^T \mathbf{\beta} + v_i + e_{ij} \quad \text{for} \quad i = 1, \dots, m \quad \text{and} \quad j \in U_i$$
(4.1)

where $v_i \stackrel{\text{ind}}{\sim} (0, \sigma_v^2)$ are the random effects and are independent of the random errors, e_{ij} , with $e_{ij} \stackrel{\text{ind}}{\sim} (0, \sigma_e^2)$. The production system includes a slight modification to the error structure of the random errors. That is, $e_{ij} \stackrel{\text{ind}}{\sim} (0, \sigma_e^2/a_{ij})$, where $a_{ij} > 0$ are positive constants that account for heteroscedasticity.

The production system computes small area estimates for means $(\overline{Y}_{ic} = \sum_{j \in U_i} c_{ij} y_{ij} / \sum_{j \in U_i} c_{ij})$ and totals $(Y_{ic} = \sum_{j \in U_i} c_{ij} \overline{Y}_i)$. The c_{ij} values are fixed positive constants known for all population units. The addition of c_{ij} was necessary to allow the use of the system by some business surveys conducted at Statistics Canada (see Rubin-Bleuer, Jang and Godbout, 2016). The available auxiliary data are either totals $\mathbf{Z}_{ic} = \sum_{j \in U_i} c_{ij} \mathbf{z}_{ij}$, or means $\overline{\mathbf{Z}}_{ic} = \sum_{j \in U_i} c_{ij} \mathbf{z}_{ij}$.

In what follows, we provide the estimators of the population means \overline{Y}_{ic} , say $\hat{\theta}_i^{\text{SAE}}$, where i = 1, ..., M. Estimates of the corresponding totals Y_{ic} , are obtained by multiplying $\hat{\theta}_i^{\text{SAE}}$ by $\sum_{j=1}^{N_i} c_{ij}$.

The design weighted sample mean of the y's and z's are respectively

$$\overline{y}_{iwc} = \left(\sum_{j \in s_i} w_{ij} c_{ij}\right)^{-1} \sum_{j \in s_i} w_{ij} c_{ij} y_{ij}$$

and

$$\overline{\mathbf{z}}_{iwc} = \left(\sum_{j \in s_i} w_{ij} c_{ij}\right)^{-1} \sum_{j \in s_i} w_{ij} c_{ij} \mathbf{z}_{ij}.$$

The model based weighted means are

$$\overline{y}_{ia} = \left(\sum_{j \in s_i} a_{ij}\right)^{-1} \left(\sum_{j \in s_i} a_{ij} y_{ij}\right)$$

and

$$\overline{\mathbf{z}}_{ia} = \left(\sum_{j \in s_i} a_{ij}\right)^{-1} \left(\sum_{j \in s_i} a_{ij} \mathbf{z}_{ij}\right).$$

Battese et al. (1988) did not include survey design weights in their procedure, thereby forsaking design consistency unless the design was self-weighting. We refer to this estimator as EBLUP $(\hat{\theta}_i^{\text{EBLUP}})$. However, EBLUP is the most efficient estimator under model (4.1), with error structure $e_{ij} \stackrel{\text{ind}}{\sim} (0, \sigma_e^2/a_{ij})$, and this is the reason that it is included in the production system.

Kott (1989), Prasad and Rao (1999), and You and Rao (2002) proposed the use of design-consistent model based estimators for the area means by including the survey weight. The You and Rao (2002) procedure was suitably modified to reflect the heteroscedastic residuals and the c_{ij} 's. The resulting Pseudo-EBLUP estimator, denoted as PEBLUP ($\hat{\theta}_i^{\text{PEBLUP}}$), was included in the production system as it is design consistent.

The EBLUP estimator is defined as

$$\hat{\theta}_{i}^{\text{EBLUP}} = \begin{cases} \hat{\gamma}_{ia} \overline{y}_{ia} + \left(\overline{\mathbf{Z}}_{ic} - \hat{\gamma}_{ia} \overline{\mathbf{z}}_{ia}\right)^{T} \hat{\boldsymbol{\beta}}^{\text{EBLUP}} & \text{if } i \in A \\ \overline{\mathbf{Z}}_{ic}^{T} \hat{\boldsymbol{\beta}}^{\text{EBLUP}} & \text{if } i \in \overline{A} \end{cases}$$

where $\hat{\gamma}_{ia} = \left(\hat{\sigma}_{v}^{2} + \hat{\sigma}_{e}^{2} / \sum_{j \in s_{i}} a_{ij}\right)^{-1} \hat{\sigma}_{v}^{2}$. The terms \overline{y}_{ia} and \overline{z}_{ia} , are the previously defined model based weighted means for y and z respectively. The regression vector $\boldsymbol{\beta}$ is estimated as

$$\hat{\boldsymbol{\beta}}^{\text{EBLUP}} = \left(\sum_{i=1}^{m} \sum_{j \in s_{i}} a_{ij} c_{ij} \left(\mathbf{z}_{ij} - \hat{\gamma}_{iac} \overline{\mathbf{z}}_{iac} \right) \mathbf{z}_{ij}^{T} \right)^{-1} \sum_{i=1}^{m} \sum_{j \in s_{i}} a_{ij} c_{ij} \left(\mathbf{z}_{ij} - \hat{\gamma}_{iac} \overline{\mathbf{z}}_{iac} \right) \mathbf{y}_{ij}.$$

The PEBLUP estimator, $\hat{\theta}_i^{\text{PEBLUP}}$, is given by

$$\hat{\theta}_{i}^{\text{PEBLUP}} = \begin{cases} \hat{\gamma}_{iwc} \,\overline{y}_{iwc} + \left(\overline{\mathbf{Z}}_{ic} - \hat{\gamma}_{iwc} \,\overline{\mathbf{Z}}_{iwc}\right)^{T} \,\hat{\boldsymbol{\beta}}^{\text{PEBLUP}} & \text{if } i \in A \\ \overline{\mathbf{Z}}_{ic}^{T} \hat{\boldsymbol{\beta}}^{\text{PEBLUP}} & \text{if } i \in \overline{A} \end{cases}$$

where $\hat{\gamma}_{iwc} = (\hat{\sigma}_v^2 + \hat{\sigma}_e^2 \delta_{iwc}^2)^{-1} (\hat{\sigma}_v^2)$, and $\delta_{iwc}^2 = \left(\sum_{j \in s_i} w_{ij} c_{ij}\right)^{-2} \left(\sum_{j \in s_i} \left(w_{ij} c_{ij}\right)^2 / a_{ij}\right)$. The terms \overline{y}_{iwc} and \overline{z}_{iwc} , are the previously defined design based weighted means for y and z respectively. The regression vector $\boldsymbol{\beta}$ is estimated as

$$\hat{\boldsymbol{\beta}}^{\text{PEBLUP}} = \left(\sum_{i=1}^{m} \sum_{j \in s_{i}} w_{ij} a_{ij} \left(\mathbf{z}_{ij} - \hat{\gamma}_{iwa} \overline{\mathbf{z}}_{iwa} \right) \mathbf{z}_{ij}^{T} \right)^{-1} \sum_{i=1}^{m} \sum_{j \in s_{i}} w_{ij} a_{ij} \left(\mathbf{z}_{ij} - \hat{\gamma}_{iwa} \overline{\mathbf{z}}_{iwa} \right) y_{ij}$$

where $\overline{\mathbf{z}}_{iwa} = \left(\sum_{j \in s_i} w_{ij} a_{ij}\right)^{-1} \sum_{j \in s_i} w_{ij} a_{ij} \mathbf{z}_{ij}, \ \hat{\gamma}_{iwa} = \left(\hat{\sigma}_v^2 + \hat{\sigma}_e^2 \delta_{iwa}^2\right)^{-1} \hat{\sigma}_v^2$ and with δ_{iwa}^2 computed as $\delta_{iwa}^2 = \left(\sum_{j \in s_i} w_{ij} a_{ij}\right)^{-2} \left(\sum_{j \in s_i} (w_{ij} a_{ij})^2 / a_{ij}\right).$

The components of variance, σ_e^2 and σ_v^2 , are estimated using the fitting-of-constants (not weighted by the survey weights) method, as given by Battese et al. (1988) or Rao (2003). The resulting estimators of σ_e^2 are always greater than or equal to zero, but the estimator of σ_v^2 may be negative. If $\hat{\sigma}_v^2 < 0$, it is set to zero, implying that there are no area effects. The associated estimated MSEs are obtained by extending You and Rao (2002) and Stukel and Rao (1997).

Note that if the sample s is selected from the universe U, the realized sampling fraction, $f_i = n_i / N_i$, could be non-negligible. For estimating a population mean, \overline{Y}_i , Rao and Molina (2015), accounted for non-negligible sampling fractions by expressing it as

$$\overline{Y}_i = f_i \overline{y}_{is} + (1 - f_i) \overline{y}_{i\overline{s}}$$

where \overline{y}_{is} is the sample mean of the *i*th sampled area and $\overline{y}_{i\overline{s}}$ is the sample mean of the non-sampled units within that area. They predicted $\overline{y}_{i\overline{s}}$ using the unit level model given by equation (4.1). Their expressions correspond to the case when $c_{ij} = 1$. This estimator was extended by Rubin-Bleuer (2014) to include the EBLUP and PEBLUP estimators for the case that c_{ij} is arbitrary. Specific details that also account for MSE estimation can be found in Estevao et al. (2015).

4.1 Benchmarking

The current production system does not have a procedure to benchmark the estimates obtained via the unit level model. However, the difference adjustment approach can be suitably modified to allow this. The EBLUP and PEBLUP estimators are of the form

$$\hat{\theta}_{i}^{\text{SAE}} = \begin{cases} \hat{\gamma}_{i}^{*} \overline{y}_{i}^{*} + (\overline{\mathbf{Z}}_{ic} - \hat{\gamma}_{i}^{*} \overline{\mathbf{z}}_{i}^{*})^{T} \hat{\boldsymbol{\beta}}^{\text{SAE}} & \text{if } i \in A \\ \overline{\mathbf{Z}}_{ic}^{T} \hat{\boldsymbol{\beta}}^{\text{SAE}} & \text{if } i \in \overline{A} \end{cases}$$

where $\hat{\gamma}_{i}^{*}$, \overline{y}_{i}^{*} , \overline{z}_{i}^{*} , and $\hat{\beta}^{\text{SAE}}$ correspond to the terms defined previously: $\hat{\gamma}_{i}^{*}$ is equal to $\hat{\gamma}_{ia}$ for EBLUP, and to $\hat{\gamma}_{iwc}$ for PEBLUP; \overline{y}_{i}^{*} is equal to \overline{y}_{ia} for EBLUP, and to \overline{y}_{iwc} for PEBLUP; \overline{z}_{i}^{*} is equal to \overline{z}_{ia} for EBLUP, and to \overline{z}_{iwc} for PEBLUP; and, $\hat{\beta}^{\text{SAE}}$ is equal to $\hat{\beta}^{\text{EBLUP}}$ for EBLUP, and to $\hat{\beta}^{\text{PEBLUP}}$ for PEBLUP.

Suppose that $\hat{\theta}_i^{\text{SAE}}$ needs to be benchmarked to $\hat{\theta}^*$. The corresponding benchmarked estimator is

$$\hat{\theta}_{i}^{\text{SAE, b}} = \begin{cases} \hat{\theta}_{i}^{\text{SAE}} + \alpha_{i} \left(\theta^{*} - \sum_{d \in A} \omega_{d} \hat{\theta}_{d}^{\text{SAE}} \right) & \text{if } i \in A \\ \overline{\mathbf{Z}}_{ic}^{T} \hat{\boldsymbol{\beta}}^{\text{SAE}} & \text{if } i \in \overline{A} \end{cases}$$

where $\alpha_i = \left(\sum_{d \in A} \omega_d^2 \tau_d\right)^{-1} (\omega_i \tau_i)$. The ω_i term is defined as follows: $\omega_i = 1$ if the benchmarking is to a total and $\omega_i = N_i / N$ if the benchmarking is for the mean. Possible choices of the τ_i 's are $\hat{\sigma}_v^2 + \hat{\sigma}_e^2 \delta_{ia}^2$, $\delta_{ia}^2 = \left(\sum_{j=1}^{n_i} a_{ij}\right)^{-1}$, for EBLUP, and $\hat{\sigma}_v^2 + \hat{\sigma}_e^2 \delta_{iwc}^2$ for PEBLUP.

4.2 Mean squared error estimation

The mean squared error estimates of the unit level estimators are based on estimating its mean squared error, given model (4.1) and error structure $e_{ij} \stackrel{\text{ind}}{\sim} (0, \sigma_e^2/a_{ij})$. Table 4.1 displays these estimated MSE's.

EstimatormseEBLUP $mse(\hat{\theta}_i^{EBLUP}) = \begin{cases} g_{1ia} + g_{2ia} + 2g_{3ia} & \text{for } i \in A \\ \overline{\mathbf{Z}}_i^T \operatorname{var}(\hat{\boldsymbol{\beta}}^{EBLUP}) \overline{\mathbf{Z}}_i + \hat{\sigma}_v^2 & \text{for } i \in \overline{A} \end{cases}$ PEBLUP $mse(\hat{\theta}_i^{PEBLUP}) = \begin{cases} g_{1iw} + g_{2iw} + 2g_{3iw} & \text{for } i \in A \\ \overline{\mathbf{Z}}_i^T \operatorname{var}(\hat{\boldsymbol{\beta}}^{PEBLUP}) \overline{\mathbf{Z}}_i + \hat{\sigma}_v^2 & \text{for } i \in \overline{A} \end{cases}$

Table 4.1MSE estimates for the unit level estimators

The various *g* terms in Table 4.1 can be interpreted in a similar way to those associated with the area level MSE's. The g_{1i} 's are denoted as g_{1ia} for EBLUP, and g_{1iw} for PEBLUP account for most of the MSE if the number of areas is large. The g_{2i} 's account for the estimation of β , and the $2g_{3i}$'s account for the estimation of σ_v^2 and σ_e^2 .

The estimated variances of $\hat{\beta}^{\text{EBLUP}}$ and $\hat{\beta}^{\text{PEBLUP}}$ are respectively given by

$$\operatorname{var}\left(\hat{\boldsymbol{\beta}}^{\mathrm{EBLUP}}\right) = \hat{\sigma}_{e}^{2} \left(\sum_{i \in A} \sum_{j \in s_{i}} a_{ij} \left(\mathbf{z}_{ij} - \hat{\gamma}_{ia} \overline{\mathbf{x}}_{ia} \right) \mathbf{z}_{ij}^{T} \right)^{-1}$$

and

$$\operatorname{var}\left(\hat{\boldsymbol{\beta}}^{\text{PEBLUP}}\right) = \hat{\sigma}_{e}^{2} \left(\sum_{i \in A} \sum_{j \in s_{i}} \boldsymbol{z}_{ij}^{*} \boldsymbol{z}_{ij}^{*T}\right)^{-1} \left(\sum_{i \in A} \sum_{j \in s_{i}} \boldsymbol{z}_{ij}^{*} \boldsymbol{z}_{ij}^{*T} \middle| a_{ij} \right) \left(\sum_{i \in A} \sum_{j \in s_{i}} \boldsymbol{z}_{ij}^{*} \boldsymbol{z}_{ij}^{*T}\right)^{-1}$$

where $\mathbf{z}_{ij}^* = w_{ij}a_{ij} \left(\mathbf{z}_{ij} - \hat{\gamma}_{iwa} \overline{\mathbf{z}}_{iwa} \right).$

The specific form of the g terms and the estimated variances can be found in Estevao et al. (2015).

5 Hierarchical Bayes (HB) method

The basic Fay-Herriot area level model includes a linear sampling model for direct survey estimates and a linear linking model for the parameters of interest. Such models are *matched* because θ_i appears as a linear function in both the sampling and linking models. There are instances when these equations are not matched such as when a function, $h(\theta_i)$, is modelled as a linear function of explanatory variables instead of θ_i . The *sampling model* and *linking model* pair is

$$\hat{\theta}_i = \theta_i + e_i \tag{5.1}$$

and

$$h(\theta_i) = \mathbf{z}_i^T \mathbf{\beta} + b_i v_i \tag{5.2}$$

where $e_i \stackrel{\text{ind}}{\sim} N(0, \psi_i)$ and $v_i \stackrel{\text{ind}}{\sim} N(0, \sigma_v^2)$.

The model pair given by (5.1) and (5.2) is referred to as an *unmatched* model. Nonlinear linking models are often needed in practice to provide a better model fit to the data. For example, if the parameter of interest is a probability or a rate within the range of 0 and 1, a linear linking model with normal random effects may not be appropriate. A linking model, in this case, could be a logistic or log-linear model. Such a model was used to adjust counts for detailed levels for the 2011 Census of Canada. A good description of what is involved to carry out such an adjustment can be found in Dick (1995) and You, Rao and Dick (2004).

The production system includes the following choices of $h(\theta_i)$

$$h(\theta_i) = \begin{cases} \theta_i & : \text{ Matched Fay-Herriot (FH) model} \\ \log(\theta_i) & : \text{ Unmatched log-linear model} \\ \log(\theta_i/(\theta_i + C_i)) & : \text{ Unmatched log census undercount model.} \end{cases}$$
(5.3)

The inclusion of $h(\theta_i) = \theta_i$ corresponds to the matched model represented by equations (3.1) and (3.2). An advantage of choosing the Hierarchical Bayes method is that the estimated σ_v^2 cannot be negative. The function $\log(\theta_i)$, where θ_i is equal to the population mean \overline{Y}_i , was used in Fay and Herriot (1979). Their context was to estimate per capita income (PCI) for small places in the United States with a population less than 1,000. The function $h(\theta_i)$, $\log(\theta_i/(\theta_i + C_i))$, was included to support the methodology to estimate the net undercoverage in Canadian Censuses. In this model, θ_i represents the number of individuals not counted in the census, while C_i is the known census count. As a result, $\theta_i/(\theta_i + C_i)$ is the proportion of individuals undercounted by the Census. The sampling variances, ψ_i , are assumed known for all the linking models represented by (5.2). The variances are assumed to be estimated for the first two functions (the matched Fay-Herriot and unmatched log-linear model) given in (5.3). If the sampling variances, ψ_i , are assumed known, then the unknown parameters in the sampling model (5.1) and the linking model (5.2) can be presented in a hierarchical Bayes (HB) framework as follows

$$\left[\hat{\theta}_{i} \mid \theta_{i}\right] \sim N\left(\theta_{i}, \psi_{i}\right), \quad i = 1, \dots, m$$

and

$$[h(\theta_i)|\beta, \sigma_v^2] \sim N(\mathbf{z}_i^T \boldsymbol{\beta}, b_i^2 \sigma_v^2)$$

If the sampling variances are unknown, they are estimated by adding

$$[d_i\widehat{\psi}_i \,|\, \psi_i] \sim \psi_i \chi_{d_i}^2$$

where $\chi^2_{d_i}$ follows a chi-square distribution with $d_i = (n_i - 1)$ degrees of freedom.

The model parameters β , σ_v^2 and ψ_i (when it is unknown) are assumed to obey prior distributions. The distributions used in the production system for β and σ_v^2 are the flat prior, $\pi(\beta) \propto 1$, and $\pi(\sigma_v^2) \propto (\sigma_v^2)^{-1/2}$. If ψ_i is estimated, the prior $\pi(\psi_i) \propto (\psi_i)^{-1/2}$ is added to the Bayesian model. These prior distributions are multiplied by the density functions of the distributions associated with the sampling and linking models. This yields a joint likelihood function in terms of the model parameters. This function is used to obtain a full conditional (posterior) distribution for each of the unknown parameters. For some of these, the resulting distribution has a tractable or well-known form. For others, the resulting distribution is a product of density functions with no known form. All HB methods involve estimation of the model parameters through repeated sampling of their respective full conditional distributions.

Markov Chain Monte Carlo (MCMC) methods are used to obtain estimates from the full conditional distribution of each parameter. Gibbs sampling is used repeatedly to sample from the full conditional distributions. The Gibbs sampling method (Gelfand and Smith, 1990) with the Metropolis-Hastings algorithm (Chib and Greenberg, 1995) are used to find the posterior means and posterior variances; see Estevao et al. (2015) for details. The various estimators of θ_i resulting from (5.3) are denoted as $\hat{\theta}_i^{HB}$.

5.1 Benchmarked HB estimator

Benchmarking of the estimators uses the *difference adjustment method* described in Section 3.2. That is, the benchmarked estimators $\hat{\theta}_{i}^{HB}$ are computed as

$$\hat{\theta}_{i}^{\text{HB}, b} = \begin{cases} \hat{\theta}_{i}^{\text{HB}} + \alpha_{i} \left(\hat{\theta}^{*} - \sum_{d \in A} \omega_{d} \hat{\theta}_{d}^{\text{HB}} \right) & \text{for } i \in A \\ \mathbf{z}_{i}^{T} \hat{\boldsymbol{\beta}} & \text{for } i \in \overline{A} \end{cases}$$

where $\alpha_i = \left(\sum_{i \in A} \omega_i^2 \left(\ddot{\psi}_i + b_i^2 \hat{\sigma}_v^{2\text{HB}} \right) \right)^{-1} \omega_i \left(\ddot{\psi}_i + b_i^2 \hat{\sigma}_v^{2\text{HB}} \right)$ for $i \in A$, and $\hat{\theta}^*$ is the benchmark value. The terms ω_i are defined as follows: $\omega_i = 1$ if the benchmarking is to a total, and $\omega_i = N_i / N$ if the

benchmarking is for the mean. The ψ_i 's are either known or unknown. The $\hat{\theta}^*$ can be a value provided by the user that represents the total or mean of the *y*-values of population *U*. The benchmarking ensures that $\sum_{i \in A \cup \overline{A}} \omega_i \hat{\theta}_i^{\text{HB}, b} = \hat{\theta}^*$.

6 Application to Labour Force Survey (LFS) data

Statistics Canada's LFS is a monthly survey with a stratified two-stage design. It is designed to produce reliable unemployment rate estimates for the 55 Employment Insurance Economic Regions (EIER) in Canada. The unemployment rate in any given area i is defined as the ratio

$$\theta_i = \frac{\sum_{j \in U_i} y_{1j}}{\sum_{j \in U_i} y_{2j}},$$

where y_{1j} is a binary variable indicating whether person j is unemployed $(y_{1j} = 1)$ or not $(y_{1j} = 0)$, and y_{2j} is a binary variable indicating whether person j is in the labour force $(y_{2j} = 1)$ or not $(y_{2j} = 0)$. The direct estimator of θ_i is the calibration composite estimator described in Fuller and Rao (2001). See also Singh, Kennedy and Wu (2001) and Gambino, Kennedy and Singh (2001). It can be written in the weighted form

$$\hat{\theta}_i = \frac{\sum_{j \in s_i} w_j y_{1j}}{\sum_{j \in s_i} w_j y_{2j}},$$

where w_i is a calibration composite weight for person *j*.

As mentioned above, the calibration composite estimator is reliable for the estimation of the unemployment rate for the 55 EIERs. There is also interest in obtaining reliable estimates for 149 areas (cities) in Canada. Among them, there are 34 Census Metropolitan Areas (CMA) and 115 Census Areas (CA). The CMAs are the largest cities in terms of population size and they usually have a large sample size as well. Some of the CAs have a very small sample size, sometimes even 0. For those CAs and other larger CAs, the sample size is not large enough to produce sufficiently reliable direct estimates of the monthly unemployment rate. Our objective was to investigate whether the Fay-Herriot model could be used to obtain monthly estimates that would be reliable enough to be published.

We constructed an auxiliary variable z_{1i} , for area *i*, given by $z_{1i} = N_i^{\text{EIB}} / N_i^{15+}$, where N_i^{EIB} is the number of employment insurance beneficiaries in area *i* and N_i^{15+} is the number of persons aged 15 years or older in area *i*. The numerator is obtained from an administrative source, whereas the denominator is a Census projection computed by Statistics Canada. We used the vector $\mathbf{z}_i = (1, z_{1i})^T$, along with $b_i = 1$, i = 1, ..., m, to obtain SAE estimates. We used May 2016 data in this investigation to allow the comparison of direct and SAE estimates with 2016 Census estimates.

Some of the 149 areas of interest had a very small sample size in the LFS: they were not used in the Fay-Herriot and smoothing models. As a rule of thumb, we excluded from the models, areas where the number of sampled persons in the labour force was smaller than 10. There were 9 such areas; among them, six had no sampled person in the labour force. Also, there were 9 other areas where the direct unemployment rate estimate, $\hat{\theta}_i$, and its direct variance estimate, $\hat{\psi}_i$, were both equal to 0. As these direct estimates were not deemed to be reliable enough, their associated areas were excluded from the models. This resulted in using only 131 areas in the models. For those areas, the small area estimates are EBLUP estimates, with the remaining 18 being synthetic estimates.

The estimator $\hat{\psi}_i$ of the direct variance ψ_i was obtained via the Rao-Wu bootstrap. The estimates of the smooth design variances were then obtained by using $\mathbf{x}_i = (1, \log(z_{1i}), \log(1 - z_{1i}), \log(N_i^{15+}))^T$. A graph of the residuals of the smoothing model, $\log(\hat{\psi}_i) - \mathbf{x}'_i \hat{\mathbf{a}}$, versus the predicted values, $\mathbf{x}'_i \hat{\mathbf{a}}$, did not reveal any obvious model misspecification. Figure 6.1 shows a graph of direct variances estimates, $\hat{\psi}_i$, versus smooth variance estimates, $\hat{\psi}_i$. The red line is the identity line. If the smoothing model is appropriate, for any value of $\hat{\psi}_i$, the average of direct variance estimates for areas around area *i* should be roughly equal to $\hat{\psi}_i$. This means that the red line should pass roughly through the middle of the points everywhere. From a quick inspection of Figure 6.1, we observe that the red line is close to the middle of the points although probably slightly above the middle due to some extreme values of $\hat{\psi}_i$. This may result in a slight overestimation of the true smooth variance $\tilde{\psi}_i = E_{mp}(\hat{\psi}_i)$. A slight overestimation is not a major issue. What has to be avoided is an underestimation of $\tilde{\psi}_i$, as it typically leads to underestimating the MSE of the SAE estimate. This would provide the user with a false impression of precision.

Overall, we were satisfied with our smoothed variance estimates. However, for areas with large sample sizes, we set $\hat{\psi}_i = \hat{\psi}_i$ as our estimate of $\hat{\psi}_i$. We assumed that direct variance estimates were stable enough when the sample size is large. As a rule of thumb, we set $\hat{\psi}_i = \hat{\psi}_i$ when the number of sampled persons in the labour force was greater than 400. This replacement occurred for 35 areas. The strategy was used to avoid possible small model biases in $\hat{\psi}_i$ for the largest areas, which could result in EBLUP estimates that become significantly different from the direct estimates. This is not a desirable property for areas with a large sample size.

The smooth variance estimates were then used to obtain small area estimates for the 149 areas of interest. Figure 6.2 shows a graph of small area and direct estimates as a function of sample size (number of sampled persons in the labour force). The small area estimates are much less volatile than direct estimates, especially for the areas with the smallest sample sizes. For the largest areas, as expected, both estimates are similar.

We first evaluated the quality of the underlying Fay-Herriot model before looking at the MSE estimates. Figure 6.3 shows the graph of direct estimates, $\hat{\theta}_i$, versus predicted values, $\mathbf{z}_i^T \hat{\boldsymbol{\beta}}$. The red line is the identity line and the blue line is a nonparametric smoothing spline curve. If the linearity assumption holds, the blue line should be close to the red line and the latter should pass roughly through the middle of the points everywhere. Figure 6.3 does not give any indication that the linearity assumption of the Fay-Herriot model is questionable.

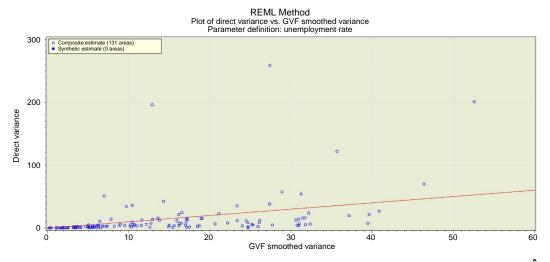


Figure 6.1 Graph of direct variance estimates, $\hat{\psi}_i$, versus smooth variance estimates, $\hat{\psi}_i$.

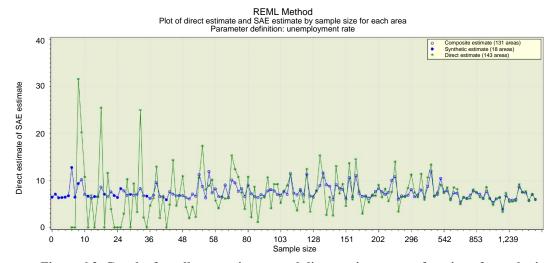


Figure 6.2 Graph of small area estimates and direct estimates as a function of sample size.

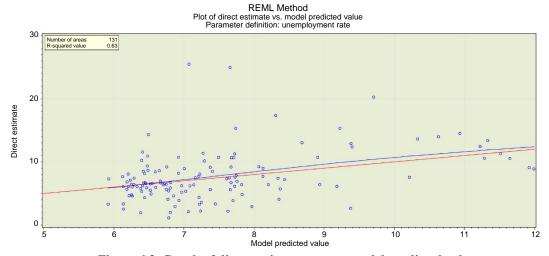


Figure 6.3 Graph of direct estimates versus model predicted values.

It is also informative to compute a measure that indicates the strength of \mathbf{z}_i for the prediction of θ_i . To this end, we developed and implemented a coefficient of determination, or R^2 value, associated with the linking model $\theta_i = \mathbf{z}_i^T \mathbf{\beta} + b_i v_i$. Note that the coefficient of determination associated with the combined model, $\hat{\theta}_i = \mathbf{z}_i^T \mathbf{\beta} + b_i v_i + e_i$, is not of interest as the objective is not the prediction of $\hat{\theta}_i$ but the prediction of θ_i . Our coefficient of determination is given by

$$R^{2} = 1 - \frac{\hat{\sigma}_{v}^{2}}{\frac{(m-q)}{(m-1)}\hat{\sigma}_{v}^{2}} + S^{2}\left(\hat{\boldsymbol{\beta}}\right)$$

where q is the dimension of \mathbf{z}_i and $S^2(\hat{\boldsymbol{\beta}})$ is the sample variance of $\mathbf{z}_i^T \hat{\boldsymbol{\beta}} / b_i$ (see equation (A.6) for the exact definition of the function $S^2(\cdot)$). The details of the derivation of the above coefficient of determination are provided in the Appendix. Figure 6.3 indicates that the R^2 value is 0.63. The linking model is thus neither weak nor extremely strong but, hopefully, strong enough to achieve efficiency gains over the direct estimator. The system also produces estimates of the parameters of the Fay-Herriot model along with their standard errors. From this output, we found out that estimates of both the intercept and slope parameters of the Fay-Herriot model were significantly different from 0 using a standard Wald test at the 0.05 significance level.

Figure 6.4 shows a graph of standardized residuals, $(\hat{\theta}_i - \mathbf{z}_i^T \hat{\boldsymbol{\beta}}) / \sqrt{b_i^2 \hat{\sigma}_v^2 + \hat{\psi}_i}$, versus standardized predicted values, $\mathbf{z}_i^T \hat{\boldsymbol{\beta}} / \sqrt{b_i^2 \hat{\sigma}_v^2 + \hat{\psi}_i}$. The red line is a horizontal line at zero and the blue line is a nonparametric smoothing spline curve. Similarly to Figure 6.3, the blue line should be close to the red line under linearity and the latter should pass roughly through the middle of the points everywhere. Again, Figure 6.4 does not indicate any obvious failure of the linearity assumption underlying the Fay-Herriot model.

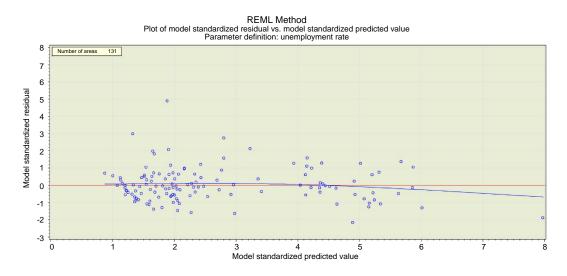


Figure 6.4 Graph of standardized residuals versus standardized predicted values.

Figure 6.5 shows a graph of squared standardized residuals versus standardized predicted values. The red line is a horizontal line at one and the blue line is again a nonparametric smoothing spline curve. This graph is used to check the homoscedasticity assumption; i.e., the assumption that the model variance σ_{ν}^2 is constant. Under homoscedasticity, the blue line should be close to the red line everywhere. The graph does not reveal any obvious presence of heteroscedasticity.

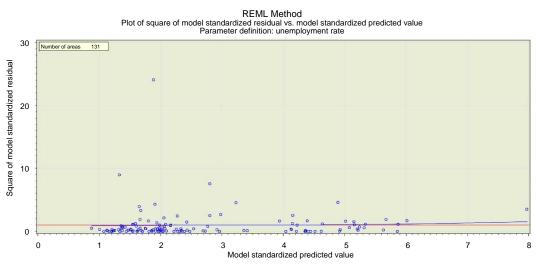


Figure 6.5 Graph of square standardized residuals versus standardized predicted values.

Figure 6.6 shows a QQ-plot of standardized residual quantiles versus standard normal quantiles. It is used to verify the normality assumption of the errors $b_i v_i$ and e_i . The graph does indicate a modest departure from normality. However, Rao and Molina (2015, page 138) argued that EBLUP estimates and their corresponding MSE estimates are generally robust to deviations from normality.

The system also computes Cook's distances to identify areas that could have a significant influence on the estimate $\hat{\beta}$. The Cook distance for area *i* is given by

$$D_i = \frac{1}{q} \left(\hat{\boldsymbol{\beta}} - \hat{\boldsymbol{\beta}}^{(-i)} \right)^T \sum_{j=1}^m \frac{\mathbf{z}_j \mathbf{z}_j^T}{b_j^2 \hat{\sigma}_v^2 + \hat{\psi}_j} \left(\hat{\boldsymbol{\beta}} - \hat{\boldsymbol{\beta}}^{(-i)} \right),$$

where $\hat{\beta}^{(-i)}$ is the estimate of β obtained after deleting area *i*. A plot of the influences D_i is provided in Figure 6.7. One area seems to have a relatively large influence compared with other areas $(D_i = 1.2851)$. This area has the largest standardized predicted value and the second largest predicted value. Its standardized residual is -1.88, which is not extreme, although not very small either. Its sample size is large (number of sampled persons in the labour force close to 500) and its smooth variance estimate, $\hat{\psi}_i$, is relatively small compared with other areas. All these reasons explain why this area was detected as being influential. In this application, we decided to keep this area in the model as its influence was not large enough to make a big difference in the SAE estimates and their corresponding MSE estimates.

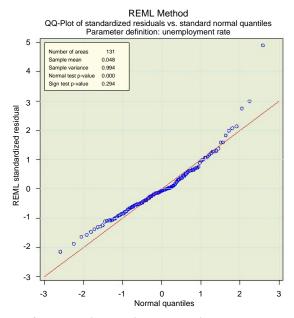
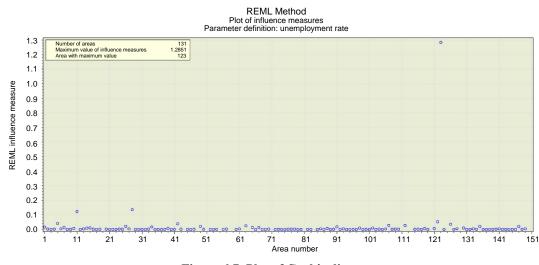
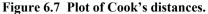


Figure 6.6 QQ-plot of standardized residual quantiles versus standard normal quantiles.





Since the Fay-Herriot model and smoothing model were both reasonable, we computed MSE estimates to evaluate the magnitude of the efficiency gains, if any, obtained by using the Fay-Herriot model. Figure 6.8 shows the estimated direct Coefficient of Variation (CV), defined as $\sqrt{\hat{\psi}_i}/\hat{\theta}_i$, and the estimated SAE Relative Root Mean Square Error (RRMSE), defined as $\sqrt{\hat{\phi}_i}/\hat{\theta}_i^{SAE}$, where $\hat{\phi}_i$ is an estimate of the MSE, $E_{mp} \left(\hat{\theta}_i^{SAE} - \theta_i\right)^2$, and $\hat{\theta}_i^{SAE}$ is the small area estimate (EBLUP or synthetic estimate) of θ_i . The sample size (number of sampled persons in the labour force) is given on the horizontal axis. The estimated direct CVs are in general much larger than the estimated SAE RRMSEs, especially for the areas with the smallest sample sizes. The estimated SAE RRMSEs are never above 20% whereas the estimated direct CV is over

300% for one area. The estimated SAE RRMSEs are also very stable as a function of the sample size unlike the erratic behavior of the estimated direct CVs. For the areas with the largest sample sizes, both estimates are very similar, as expected. This indicates that SAE methods can lead to a substantial increase of precision over direct estimation methods, particularly for the smallest areas.

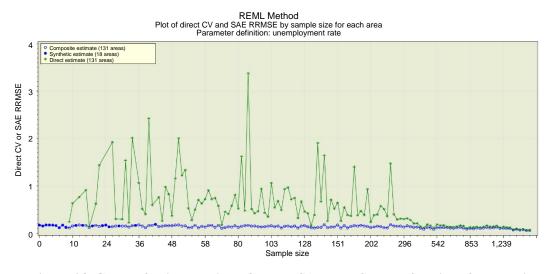


Figure 6.8 Graph of estimated direct CVs and SAE RRMSEs as a function of sample size.

For the month of May 2016, we had the luxury of having a very reliable source for the estimation of the unemployment rates: the 2016 long form Census administered to roughly one-fourth of the households throughout Canada. The Census sample size is much larger than the LFS sample size in all the areas of interest. Therefore, we used the 2016 Census direct estimates, denoted by $\hat{\theta}_i^{\text{Census}}$, as a gold standard for evaluating the accuracy of both the LFS direct estimates and SAE estimates. We computed Absolute Relative Differences (ARD) between LFS direct estimates and Census estimates, $|\hat{\theta}_i - \hat{\theta}_i^{\text{Census}}| / \hat{\theta}_i^{\text{Census}}$, as well as ARDs between SAE estimates and Census estimates, $|\hat{\theta}_i^{\text{SAE}} - \hat{\theta}_i^{\text{Census}}| / \hat{\theta}_i^{\text{Census}}$. These ARDs were then averaged within 5 different homogeneous subgroups with respect to sample size. Table 6.1 summarizes the results.

Table 6.1
Average ARD of SAE estimates and LFS direct estimates expressed in percentage

Sample size	Average ARD between LFS direct estimates and Census estimates	Average ARD between SAE estimates and Census estimates	Average ARD between HB estimates and Census estimates		
28 smallest areas	70.4%	17.7%	18.3%		
Next 28 smallest areas	38.7%	18.9%	19.0%		
Next 28 smallest areas	26.2%	13.8%	14.1%		
Next 28 smallest areas	20.9%	12.7%	13.0%		
28 largest areas	13.2%	10.2%	10.3%		
Overall	33.9%	14.7%	14.9%		

Note: Out of the 149 areas of interest in the LFS, 9 were excluded from this table: six where the LFS number of sampled persons in the labour force was 0 and three that were no longer in the list of CMAs /CAs after the 2016 Census.

As expected, the ARD between the LFS and Census direct estimates decreases as the sample size increases. This may suggest that the conceptual differences between these two surveys and nonsampling errors are reasonably small compared with the sampling error, especially for the smallest areas where the sampling error may be the main contributor to the ARD. The SAE estimates are much closer to the Census estimates than the LFS direct estimates, particularly for the smallest areas where improvement is most needed. This confirms that our underlying models are reasonable in this application.

For comparison purposes, we also computed HB estimates, $\hat{\theta}_i^{\text{HB}}$, based on the matched Fay-Herriot model with the noninformative priors for β , σ_v^2 and $\tilde{\psi}_i$ provided in Section 5. We then computed ARDs between HB estimates and Census estimates, $|\hat{\theta}_i^{\text{HB}} - \hat{\theta}_i^{\text{Census}}| / \hat{\theta}_i^{\text{Census}}$. Results are given in the last column of Table 6.1. The averaged ARDs of the HB estimates are close to those of the EBLUP estimates.

7 Conclusion

A frequent demand from users of data from NSOs is for more granularity for use in planning and policy research purposes. NSOs can no longer simply increase the sample sizes of their surveys to obtain reliable estimates at the requested level of detail. Reasons for this include the high costs of doing so, response burden concerns, as well as the difficult task of obtaining responses from sampled units. An alternative being investigated by many NSOs is the use of small area estimation techniques that provide a way to address the demand for more granular data. With this in mind, Statistics Canada began the development of an SAE production system in the early 2000s and now have such a system available to their statistical programs. The production system handles area and unit level models, with multiple options such as different methods to estimate the variance components, different linking models and both the EBLUP and HB estimation methods. It is currently being used to produce experimental estimates for several Statistics Canada statistical programs and it is expected that the first published small area estimates will be available in 2019.

As it was mentioned in the introduction, the only existing software in 2006 that would produce small area estimates and their associated mean squared estimates was sponsored by the EURAREA (2004) project. The current production system developed at Statistics Canada is written in SAS, with its methodology closely following Rao (2003) and includes some recent advances. As it stands, it satisfies the existing requirements for small area estimation at Statistics Canada. However, as the use of small area estimation becomes more common within Statistics Canada, there will be a need to add functionality to the system to meet this demand. The recent book authored by Rao and Molina (2015) provides an idea of how much development has taken place in small area estimation during recent years. The incorporation of all this development into the production system would be extremely time consuming, expensive, and may not be directly applicable to the needs of Statistics Canada. It, therefore, follows that options other than programming these new functionalities in the current SAS production system should be considered. One option would be to investigate how packages developed elsewhere, such as those written in R, can be integrated into it. Notable packages written in R include *sae* (Molina and Marhuenda, 2015), *mme*

(Lopez-Vizcaino, Lombardia and Morales, 2014), *saery* (Esteban, Morales and Perez, 2014) and *sae2* (Fay and Diallo, 2015). These packages include small area procedures that are not in the present system such as multinomial linear mixed models, area level models with time effects and time series area level models supporting univariate and multivariate applications. The existing SAS production system meets the needs of Statistics Canada at this point in time, and there are no concrete plans to add functionality to it.

Acknowledgements

We would like to thank the reviewers for their comments and suggestions that led to improvements in this paper.

Appendix

Justification of the coefficient of determination

In order to determine a coefficient of determination associated with the linking model, $\theta_i = \mathbf{z}_i^T \mathbf{\beta} + b_i v_i$, we first rewrite it as

$$\tilde{\theta}_i = \tilde{\mathbf{z}}_i^T \mathbf{\beta} + v_i$$

where $\tilde{\theta}_i = \theta_i / b_i$ and $\tilde{\mathbf{z}}_i = \mathbf{z}_i / b_i$. We assume that an intercept is implicitly or explicitly included in $\tilde{\mathbf{z}}_i$; i.e., there exists a vector $\boldsymbol{\lambda}$ such that $\boldsymbol{\lambda}^T \tilde{\mathbf{z}}_i = 1$. In other words, we assume that there exists a vector $\boldsymbol{\lambda}$ such that $b_i = \boldsymbol{\lambda}^T \mathbf{z}_i$. If $\tilde{\theta}_i$, i = 1, ..., m, were known, we could estimate the unknown vector of model parameters $\boldsymbol{\beta}$ by the least squares estimator

$$\hat{\boldsymbol{\beta}}_* = \left(\sum_{i=1}^m \tilde{\boldsymbol{z}}_i \tilde{\boldsymbol{z}}_i^T\right)^{-1} \sum_{i=1}^m \tilde{\boldsymbol{z}}_i \tilde{\boldsymbol{\theta}}_i$$

and the unknown model variance σ_v^2 by the unbiased estimator

$$\hat{\sigma}_{v^*}^2 = \frac{\sum_{i=1}^m \left(\tilde{\theta}_i - \tilde{\mathbf{z}}_i^T \hat{\boldsymbol{\beta}}_*\right)^2}{m-q}.$$

The well-known adjusted coefficient of determination is

$$R_{\rm ideal}^{2} = 1 - \frac{\hat{\sigma}_{\nu^{*}}^{2}}{\left(m-1\right)^{-1} \sum_{i=1}^{m} \left(\tilde{\theta}_{i} - \overline{\tilde{\theta}}\right)^{2}},\tag{A.1}$$

where $\overline{\hat{\theta}} = m^{-1} \sum_{i=1}^{m} \hat{\theta}_i$. It is an ideal coefficient of determination because it cannot be computed (since $\hat{\theta}_i$ is unknown) but this is the target we would like to estimate. Simply replacing θ_i with $\hat{\theta}_i$ does not solve the problem as $\hat{\theta}_i$ reflects the combined model and not just the linking model. The resulting coefficient of

determination would typically be too small. To obtain a better estimate of R_{ideal}^2 , we first decompose $\sum_{i=1}^{m} (\tilde{\theta}_i - \overline{\tilde{\theta}})^2$ as

$$\sum_{i=1}^{m} \left(\tilde{\theta}_{i} - \overline{\tilde{\theta}} \right)^{2} = \sum_{i=1}^{m} \left(\tilde{\theta}_{i} - \tilde{\mathbf{z}}_{i}^{T} \hat{\boldsymbol{\beta}}_{*} \right)^{2} + \sum_{i=1}^{m} \left(\tilde{\mathbf{z}}_{i}^{T} \hat{\boldsymbol{\beta}}_{*} - \overline{\tilde{\theta}} \right)^{2} + 2 \sum_{i=1}^{m} \left(\tilde{\theta}_{i} - \tilde{\mathbf{z}}_{i}^{T} \hat{\boldsymbol{\beta}}_{*} \right) \left(\tilde{\mathbf{z}}_{i}^{T} \hat{\boldsymbol{\beta}}_{*} - \overline{\tilde{\theta}} \right).$$
(A.2)

Assuming that an intercept is implicitly or explicitly included in \tilde{z}_i and from the expression for β_* , we have that

$$\sum_{i=1}^{m} \left(\tilde{\theta}_{i} - \tilde{\mathbf{z}}_{i}^{T} \hat{\boldsymbol{\beta}}_{*} \right) \tilde{\mathbf{z}}_{i} = \mathbf{0}$$
(A.3)

and

$$\sum_{i=1}^{m} \left(\tilde{\theta}_i - \tilde{\mathbf{z}}_i^T \hat{\boldsymbol{\beta}}_* \right) = 0.$$
 (A.4)

From (A.4), we can rewrite $\overline{\tilde{\theta}}$ as $\overline{\tilde{\theta}} = \overline{\tilde{z}}^T \hat{\beta}_*$, where $\overline{\tilde{z}} = m^{-1} \sum_{i=1}^m \tilde{z}_i$. As a result, the cross product term in (A.2) vanishes and equation (A.2) reduces to

$$\sum_{i=1}^{m} \left(\tilde{\theta}_{i} - \overline{\tilde{\theta}} \right)^{2} = \sum_{i=1}^{m} \left(\tilde{\theta}_{i} - \tilde{\mathbf{z}}_{i}^{T} \hat{\mathbf{\beta}}_{*} \right)^{2} + \sum_{i=1}^{m} \left(\tilde{\mathbf{z}}_{i}^{T} \hat{\mathbf{\beta}}_{*} - \overline{\tilde{\mathbf{z}}}^{T} \hat{\mathbf{\beta}}_{*} \right)^{2}$$

$$= (m-q) \hat{\sigma}_{v^{*}}^{2} + (m-1) S^{2} \left(\hat{\mathbf{\beta}}_{*} \right),$$
(A.5)

where

$$S^{2}\left(\hat{\boldsymbol{\beta}}_{*}\right) = \frac{\sum_{i=1}^{m} \left(\tilde{\boldsymbol{z}}_{i}^{T}\hat{\boldsymbol{\beta}}_{*} - \overline{\tilde{\boldsymbol{z}}}^{T}\hat{\boldsymbol{\beta}}_{*}\right)^{2}}{m-1}.$$
(A.6)

From (A.5), it follows that the ideal coefficient of determination (A.1) can be rewritten as

$$R_{\text{ideal}}^{2} = 1 - \frac{\hat{\sigma}_{\nu^{*}}^{2}}{\frac{(m-q)}{(m-1)}\hat{\sigma}_{\nu^{*}}^{2} + S^{2}\left(\hat{\beta}_{*}\right)} \equiv f(\hat{\beta}_{*}, \hat{\sigma}_{\nu^{*}}^{2}).$$
(A.7)

The only unknown quantities in (A.7) are $\hat{\boldsymbol{\beta}}_*$ and $\hat{\sigma}_{\nu^*}^2$. A computable coefficient of determination can thus be obtained by replacing $\hat{\boldsymbol{\beta}}_*$ and $\hat{\sigma}_{\nu^*}^2$ in (A.7) with $\hat{\boldsymbol{\beta}}$ and $\hat{\sigma}_{\nu}^2$, the consistent estimators of $\boldsymbol{\beta}$ and σ_{ν}^2 implemented in the SAE system and described in Section 3. The resulting coefficient of determination can be expressed as $R^2 = f(\hat{\boldsymbol{\beta}}, \hat{\sigma}_{\nu}^2)$, with the function $f(\cdot, \cdot)$ defined in (A.7), and is a consistent estimator of the ideal coefficient of determination R_{ideal}^2 .

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Weighted censored quantile regression

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Abstract

In this paper, we make use of auxiliary information to improve the efficiency of the estimates of the censored quantile regression parameters. Utilizing the information available from previous studies, we computed empirical likelihood probabilities as weights and proposed weighted censored quantile regression. Theoretical properties of the proposed method are derived. Our simulation studies shown that our proposed method has advantages compared to standard censored quantile regression.

Key Words: Empirical Likelihood; Right censoring; Kaplan-Meier Estimator.

1 Introduction

In quantile regression (Koenker, 2005), the conditional quantiles of the response variable for a given set of predictor variables are modelled. The regression parameters are estimated by minimizing a check loss function at a specific quantile, τ , instead of the square loss function as in the standard linear regression. A quantile regression model based on properly selected quantiles could provide a global assessment of the covariate effects on the response, which is often ignored by the standard linear regression model. Recently, censored quantile regression has been studied extensively. Powell (1984) introduced the least absolute deviation (LAD) estimator, also called the median regression model for the left censored survival data, using the censored Tobit model (Tobin, 1958). Powell (1986) generalized the LAD estimation to any quantile.

Portnoy (2003) introduced a censored quantile regression model under random censoring as a generalization of the Kaplan-Meier estimator recursively using the Kaplan-Meier estimator (Kaplan and Meier, 1958). Peng and Huang (2008) developed a censored quantile regression model based on the Nelson-Aalen estimator using counting processes and martingale theory. In survival analysis setup, for the *i*th (i = 1, 2, ..., n) subject, let T_i be the logarithm of the failure time, C_i the logarithm of right censoring time, \mathbf{X}_i the *p*-vector covariate and let $Y_i = \min(T_i, C_i)$ be the logarithm of the survival time. For a given quantile, τ , the regression coefficients, $\boldsymbol{\beta}(\tau)$, can be estimated as

$$\hat{\boldsymbol{\beta}}(\tau) = \arg\min_{\boldsymbol{\beta} \in \mathfrak{R}^p} \sum_{i=1}^n \rho_{\tau} \left(Y_i - \min\{C_i, \mathbf{X}_i^{\mathsf{T}} \boldsymbol{\beta}\} \right), \tag{1.1}$$

where $\rho_{\tau}(u) = u[\tau - \mathbb{I}(u < 0)]$, is the check loss function.

In many studies, we may have some information about the target population from previous studies. This is common in survey sampling since surveys are carried out repeatedly with similar objectives. For example, in survey sampling, information about the population mean and variance could be available from previous surveys or records. The information of the parameters as well as type of relationship, distributional

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assumptions, etc. also could be considered as auxiliary information available for analysis. The auxiliary information could be effectively used to improve the efficiency of the statistical inference (Kuk and Mak, 1989; Rao, Kovar and Mantel, 1990; Chen and Qin, 1993). The idea used in this paper can be easily extendable in survey sampling to arrive efficient parameter estimates by making use of the information available from previous surveys.

Consider a known relationship between the survival time, Y (or the failure time, T) and a set of covariates **X**, as $Y = f(\mathbf{X}; \mathbf{\theta})$, where $\mathbf{\theta}$ is the parameter of interest. The knowledge about this relationship can be treated as auxiliary information. In a more general case, the auxiliary information can be expressed as $E\{g(\mathbf{Z}; \mathbf{\theta})\} = 0$ for some d-dimensional parameter, $\mathbf{\theta} \in \mathbb{R}^d$, where **Z** is the observed data from the present study and $g(\mathbf{Z}; \mathbf{\theta}) \in \mathbb{R}^q$, some function with $q \ge d$. The parameter, $\mathbf{\theta}$ could be unknown, but can be estimated using the information available from previous studies.

Chen and Qin (1993) introduced the use of auxiliary information to improve the efficiency of estimators in the context of survey sampling using empirical likelihood (Owen, 1988, 2001). Li and Wang (2003) accommodated the auxiliary information to the censored linear regression model using empirical likelihood by defining a synthetic variable (Koul, Susarla and Ryzin, 1981). Fang, Li, Lu and Qin (2013) proposed the effective use of auxiliary information in the linear regression model with right censored data using empirical likelihood, by utilizing the Buckley-James (Buckley and James, 1979) estimating equation. Tang and Leng (2012) introduced an empirical likelihood based linear quantile regression model using auxiliary information.

In this paper, we propose an empirical likelihood (EL) based approach to accommodate auxiliary information to the censored quantile regression. EL is a non-parametric likelihood approach proposed by Owen (1988, 2001), which has similar properties of parametric likelihood. We utilize the EL based data driven probabilities as the weights by using the estimating function, $g(\mathbf{Z}; \boldsymbol{\theta})$ and incorporate those weights into the censored quantile regression model. The resulted weighted censored quantile regression parameter $\beta(\tau)$ can be estimated as

$$\hat{\boldsymbol{\beta}}(\tau) = \arg\min_{\boldsymbol{\beta} \in \mathfrak{R}^{p}} \sum_{i=1}^{n} \omega_{i} \rho_{\tau} \left(Y_{i} - \min\left\{ C_{i}, \mathbf{X}_{i}^{\mathsf{T}} \boldsymbol{\beta} \right\} \right),$$
(1.2)

where ω_i 's are the weights. We propose to use the EL based data driven probabilities as the weights. Our simulation results show that the EL based weighted censored quantile regression performs more efficiently than the standard linear censored quantile regression.

The rest of the paper is organized as follows. In Section 2, we present the estimation procedure of the EL based data driven probabilities. In Section 3, we introduce the EL based weighted censored quantile regression and investigate the asymptotic properties of the estimators. In Section 4, performance analysis of the proposed method is conducted using the simulations. The application to the north central cancer treatment lung cancer data is also presented as an illustration. Our conclusions are given in Section 5.

2 Estimation of weights using empirical likelihood

We develop a method that converts the auxiliary information to the EL based data driven probabilities, which are further used in the weighted censored quantile regression as the weights. Qin and Lawless (1994) developed the EL approach based on a set of estimating equations. Let $\{\mathbf{Z}_i\}_{i=1}^n$ be the observed data and the available auxiliary information is represented by the estimating function $g(\mathbf{Z}_i; \boldsymbol{\theta})$ with parameter, $\boldsymbol{\theta}$ which is known. Then, the maximum empirical likelihood is given by

$$L_{\rm EL}(\mathbf{\theta}) = \sup\left\{\prod_{i=1}^{n} P_i: P_i \ge 0, \ \sum_{i=1}^{n} P_i = 1, \ \sum_{i=1}^{n} P_i g\left(\mathbf{Z}_i; \mathbf{\theta}\right) = 0\right\},\tag{2.1}$$

where $P_i = \Pr(Z_i = z_i)$ and $\boldsymbol{\theta}$ is the parameter in the auxiliary information which can be assumed to be known. The parameter, $\boldsymbol{\theta}$ could be any parametric information available from the previous studies which has an influence on the model parameter, $\boldsymbol{\beta}(\tau)$. For a given $g(\mathbf{Z}_i; \boldsymbol{\theta}), \boldsymbol{\theta}$ should satisfy $E\{g(\mathbf{Z}_i; \boldsymbol{\theta})\} = 0$ to avoid the non-existence of solutions due to convex hull issues. This is the scenario for when zero is not an inner point of the convex hull of the $g(\mathbf{Z}_i; \boldsymbol{\theta}), i = 1, 2, ..., n$, which will fail to provide positive P_i 's. For a given value of $\boldsymbol{\theta} = \boldsymbol{\theta}_0$, using the Lagrange multiplier method, $L_{\text{EL}}(\boldsymbol{\theta}_0)$ attains its maximum at

$$P_i\left(\boldsymbol{\theta}_0\right) = \frac{1}{n\left\{1 + \lambda_{\theta_0}^{\mathsf{T}} g\left(\mathbf{Z}_i; \boldsymbol{\theta}_0\right)\right\}}, \quad i = 1, 2, \dots, n.$$

$$(2.2)$$

The Lagrange multiplier, λ_{θ_0} is the solution to the equation

$$\sum_{i=1}^{n} \frac{g\left(\mathbf{Z}_{i}; \mathbf{\theta}_{0}\right)}{n\left\{1 + \lambda_{\theta_{0}}^{\mathsf{T}} g\left(\mathbf{Z}_{i}; \mathbf{\theta}_{0}\right)\right\}} = 0$$

The estimated $P_i(\cdot)$'s are used as the weights (ω_i) in (1.2) for the weighted censored quantile regression. In some cases, $\boldsymbol{\theta}$ may not be available and in such situations, we can use an estimate of $\boldsymbol{\theta}$, say $\hat{\boldsymbol{\theta}}_A$ obtained from previous studies. Chen and Qin (1993) showed that for a random sample, Y_i , and $P_i(\cdot)$'s are estimated using (2.2), $\tilde{F}_n(y) = \sum_{i=1}^n P_i \mathbb{I}(Y_i \leq y)$ has smaller variance than the empirical distribution function, $\hat{F}_n(y) = \frac{1}{n} \sum_{i=1}^n \mathbb{I}(Y_i \leq y)$. As a result, with Bahadur representation (Bahadur, 1966), for a given τ (0 < τ < 1), the quantile estimate, $\tilde{F}_n^{-1}(\tau)$ has smaller variance than $\hat{F}_n^{-1}(\tau)$ (See Kuk and Mak, 1989; Rao et al., 1990). Hence our proposed method is expected to be more efficient than the ordinary censored quantile regression.

3 Estimation of weighted censored quantile regression parameters

Define the distribution function of T_i conditional on the *p*-vector covariate, \mathbf{X}_i as $F_{T_i}(t | \mathbf{X}_i) = \Pr(T_i \le t | \mathbf{X}_i)$. Let $\Lambda_{T_i}(t | \mathbf{X}_i) = -\log\{1 - \Pr(T_i \le t | \mathbf{X}_i)\}, N_i(t) = \mathbb{I}(Y_i \le t, \delta_i = 1), \text{ and } M_i(t) = N_i(t) - \Lambda_{T_i}(t\Lambda Y_i | \mathbf{X}_i)$. Here $\Lambda_{T_i}(\cdot | \mathbf{X}_i)$ is the cumulative hazard function conditional on $\mathbf{X}_i, N_i(t)$ is the counting process and $M_i(t)$ is the martingale process associated with $N_i(t)$ (Fleming and Harrington, 2011). We consider an extension of censored quantile regression estimation procedure proposed by Peng and Huang (2008), incorporating the P_i 's as known weights arrived based on the auxiliary information

available through the known parameter θ . Note that θ and $\beta(\tau)$ are distinct parameters and estimating function $g(z; \theta)$ used for computing P_i 's are different from the estimating functions used for quantile regression parameters in (1.1). Since P_i 's are independent of $\beta(\tau)$, $E\{P_iM_i(t) | \mathbf{X}_i\} = \mathbf{0}$ (by the martingale property) for $t \ge 0$, we have

$$E\left\{\sqrt{n}\sum_{i=1}^{n}P_{i}\mathbf{X}_{i}\left(N_{i}\left(e^{\mathbf{X}_{i}^{\mathsf{T}}\boldsymbol{\beta}_{0}(\tau)}\right)-\Lambda_{T}\left[e^{\mathbf{X}_{i}^{\mathsf{T}}\boldsymbol{\beta}_{0}(\tau)}\wedge Y_{i}\mid\mathbf{X}_{i}\right]\right\}=\mathbf{0},$$
(3.1)

where $\boldsymbol{\beta}_0(\tau)$ denotes the true $\boldsymbol{\beta}(\tau)$, in (1.2) for a given quantile, τ .

Since $\Lambda_{T_i}(\cdot | \mathbf{X}_i)$, i = 1, 2, ..., n are unknown functions, Peng and Huang (2008) derived the relationship between $\Lambda_T [e^{\mathbf{X}_i^{\mathsf{T}} \boldsymbol{\beta}_0(\tau)} \wedge Y_i | \mathbf{X}_i]$ and $\boldsymbol{\beta}_0(\tau)$ to use (3.1) to estimate $\boldsymbol{\beta}_0(\tau)$. Using the fact that $F_{\mathfrak{F}}[e^{\mathbf{X}_i^{\mathsf{T}} \boldsymbol{\beta}_0(u)} | \mathbf{X}_i] = \tau$ and utilizing the monotonicity of $X_i^{\mathsf{T}} \boldsymbol{\beta}_0(\tau)$ in τ , they showed that $\Lambda_T [e^{\mathbf{X}_i^{\mathsf{T}} \boldsymbol{\beta}_0(\tau)} \wedge Y_i | \mathbf{X}_i] = \int_0^{\tau} \mathbb{I}[Y_i \ge e^{\mathbf{X}_i^{\mathsf{T}} \boldsymbol{\beta}(u)}] dH(u)$, where $H(u) = -\log(1-u)$ for $0 \le u < 1$.

So, our weighted censored quantile regression estimating equation is

$$\sqrt{n}S_n\left(\boldsymbol{\beta},\,\tau\right) = \mathbf{0},\tag{3.2}$$

where

$$S_n\left(\boldsymbol{\beta},\,\tau\right) = \sum_{i=1}^n P_i \mathbf{X}_i \left\{ \mathbb{N}_i \left(e^{\mathbf{X}_i^{\mathsf{T}}\boldsymbol{\beta}(\tau)} \right) - \int_0^\tau \mathbb{I}\left[Y_i \ge e^{\mathbf{X}_i^{\mathsf{T}}\boldsymbol{\beta}(u)} \right] dH\left(u\right) \right\}.$$

Here P_i 's are defined in (2.2). Let $s(\boldsymbol{\beta}, \tau) = E\{S_n(\boldsymbol{\beta}, \tau)\}\)$ and the martingale property of $\mathbb{M}(\cdot)$ gives $s(\boldsymbol{\beta}_0, \tau) = \mathbf{0}$. For a particular quantile, τ_k and an estimator of $\boldsymbol{\beta}_0(\tau_k)$, $\hat{\boldsymbol{\beta}}(\tau_k)$ is a right-continuous step function which jumps only on a grid, $\mathbb{S}_L = \{0 = \tau_0 < \tau_1 < ... < \tau_L = \tau_U < 1\}$. Here *L* depends on the sample size, *n*. The size of \mathbb{S}_L is defined as $\|\mathbb{S}_L\| = \max_k (\tau_k - \tau_{k-1})$.

For different quantiles, $\tau_0, \tau_1, ..., \tau_L$ ($0 = \tau_0 < \tau_1 < ... < \tau_L < 1$), based on (3.2), we can obtain $\hat{\boldsymbol{\beta}}(\tau_k)(k = 1, 2, ..., L)$ by recursively solving the following monotone estimating equation for $\boldsymbol{\beta}(\tau_k)$:

$$\sqrt{n}\sum_{i=1}^{n}P_{i}\mathbf{X}_{i}\left\{\mathbb{N}_{i}\left(e^{\mathbf{X}_{i}^{\mathsf{T}}\boldsymbol{\beta}(\tau_{k})}\right)-\sum_{r=0}^{k-1}\mathbb{I}\left[Y_{i}\geq e^{\mathbf{X}_{i}^{\mathsf{T}}\boldsymbol{\hat{\beta}}(\tau_{r})}\right]\left\{H\left(\tau_{r+1}\right)-H\left(\tau_{r}\right)\right\}\right\}=\mathbf{0}.$$
(3.3)

We define the estimators, $\hat{\boldsymbol{\beta}}(\tau_k)$ as the generalized solutions (Fygenson and Ritov, 1994) because equation (3.3) is not continuous and the solution may not be unique.

3.1 Asymptotic theory

We derived the asymptotic properties of the EL based weighted censored quantile regression estimators using the approach of Peng and Huang (2008). Now we prove the uniform consistency and weak Gaussian convergence of the proposed weighted censored quantile regression estimator, $\hat{\beta}(\cdot)$.

Define $F(t | \mathbf{X}) = \Pr(Y \le t | \mathbf{X}), \overline{F}(t | \mathbf{X}) = \Pr(Y > t | \mathbf{X}), \widetilde{F}(t | \mathbf{X}) = \Pr(Y \le t, \delta = 1 | \mathbf{X}), \overline{f}(y | \mathbf{X}) = -f(y | \mathbf{X}) = -dF(y | \mathbf{X})/dy$ and $\widetilde{f}(y | \mathbf{X}) = d\widetilde{F}(y | \mathbf{X})/dy$. (For a vector $h, h^{\otimes 2} = hh^T, h^{(l)} = l^{\text{th}}$ component of h, ||h|| is the Euclidean norm of h.)

Define $\mathbf{W}_i = \lambda_{\theta_0}^{\mathsf{T}} g(\mathbf{Z}_i; \boldsymbol{\theta}_0) \mathbf{X}_i, i = 1, 2, ..., n \text{ as a } p$ -vector.

Regularity conditions:

- R.1: The observations, \mathbf{Z}_i , i = 1, 2, ..., n are iid observations from some distribution. Without loss of generality, we assume that $(Y_i, \delta_i, \mathbf{X}_i^{\mathsf{T}})^{\mathsf{T}} \subset \mathbf{Z}_i$ for all i = 1, 2, ..., n.
- R.2: There exists $\boldsymbol{\theta}_0$ such that $E\{g(\mathbf{Z}_i; \boldsymbol{\theta}_0)\} = 0$, the matrix $\boldsymbol{\Sigma}(\boldsymbol{\theta}_0) = E\{g(\mathbf{Z}_i; \boldsymbol{\theta}_0) g(\mathbf{Z}_i; \boldsymbol{\theta}_0)^{\mathsf{T}}\}$ is positive definite, $\frac{\partial g(\mathbf{z}; \boldsymbol{\theta})}{\partial \boldsymbol{\theta}}$ is continuous in the neighborhood of $\boldsymbol{\theta}_0$. The matrix $E\{\frac{\partial g(\mathbf{Z}; \boldsymbol{\theta})}{\partial \boldsymbol{\theta}}\}$ is of full rank.
- **R.3**: There exist functions $H_{lj}(\mathbf{z})$ such that for $\boldsymbol{\theta}$ in the neighborhood of $\boldsymbol{\theta}_0$, $\left|\frac{\partial g_l(\mathbf{z};\boldsymbol{\theta})}{\partial \theta_j}\right| \le H_{lj}(\mathbf{z})$, where for a constant C, $E\left\{H_{lj}^2(\mathbf{Z})\right\} \le C < \infty$ for l = 1, ..., q and j = 1, ..., d.
- **R.4:** $\sup \| \mathbf{X}_i \| < \infty$ and $\sup \| \mathbf{X}_i \mathbf{Y}_i \| < \infty$.
- R.5: (a) Each component of $E[\mathbf{XN}(e^{\mathbf{X}^{\mathsf{T}}\boldsymbol{\beta}_{0}(\tau)})]$ is a Lipschitz function of τ .
 - (b) $\tilde{f}(t | \mathbf{x})$ and $f(t | \mathbf{x})$ are bounded above uniformly in t and \mathbf{x} .
- R.6: (a) $\tilde{f}(e^{\mathbf{X}^{\mathsf{T}_{\mathsf{b}}}}|\mathbf{X}) > 0$ for all $\mathbf{b} \in \mathfrak{B}(d_0)$, where $\mathfrak{B}(d) = \left\{ \mathbf{b} \in \mathfrak{R}^p : \inf_{\tau \in (0, \tau_U)} \| \mathbf{\mu}(\mathbf{b}) \mathbf{\mu} \{ \boldsymbol{\beta}_0(\tau) \} \| \le d \right\}$ for d > 0, and $\mathbf{\mu}(\mathbf{b}) = E[\mathbf{X}\mathbb{N}(e^{\mathbf{X}^{\mathsf{T}_{\mathsf{b}}}})]$, is a neighbourhood containing $\{ \boldsymbol{\beta}_0(\tau), \tau \in (0, \tau_U) \}$.
 - (b) To have the positive definiteness, $E\{\mathbf{X}^{\otimes 2}\} > 0$.
 - (c) Each component of $E\left[\mathbf{X}^{\otimes 2}\overline{f}\left(e^{\mathbf{X}^{\mathsf{T}}\mathbf{b}} \mid \mathbf{X}\right)e^{\mathbf{X}^{\mathsf{T}}\mathbf{b}}\right] \times \left(E\left[\mathbf{X}^{\otimes 2}\widetilde{f}\left(e^{\mathbf{X}^{\mathsf{T}}\mathbf{b}} \mid \mathbf{X}\right)e^{\mathbf{X}^{\mathsf{T}}\mathbf{b}}\right]\right)^{-1}$ is uniformly bounded in $\mathbf{b} \in \mathfrak{B}(d_{0})$; $\mathfrak{B}(d_{0})$.
- R.7: For any $v \in (0, \tau_U)$, $\inf_{\tau \in [v, \tau_U]} \operatorname{eigmin} E\left[\mathbf{X}^{\otimes 2} \tilde{f}\left(e^{\mathbf{X}^{\mathsf{T}} \boldsymbol{\beta}_0(\tau)} \mid \mathbf{X}\right) e^{\mathbf{X}^{\mathsf{T}} \boldsymbol{\beta}_0(\tau)}\right] > 0$, where $\operatorname{eigmin}(\cdot)$ denotes the minimum eigenvalue of a matrix.

Theorem 1. Assuming that the regularity conditions R.1-R.7 hold, if $\lim_{n\to\infty} \|\mathbb{S}_L\| = 0$, then $\sup_{\tau\in[\nu,\tau_U]} \|\hat{\boldsymbol{\beta}}(\tau) - \boldsymbol{\beta}_0(\tau)\| \to_p 0$, where $0 < \nu < \tau_U$.

Theorem 2. Assuming that the regularity conditions R.1-R.7 hold, if $\lim_{n\to\infty} n^{1/2} \| S_L \| = 0$, then $n^{1/2} \{ \hat{\beta}(\tau) - \beta_0(\tau) \}$ weakly converges to a zero-mean Gaussian process for $\tau \in [v, \tau_U]$, where $0 < v < \tau_U$.

To prove Theorems 1 and 2, we need to show that $\max_{1 \le i \le n} |\lambda_{\theta_0}^{\top} g(\mathbf{Z}_i; \boldsymbol{\theta}_0)| = o_p(1)$. We consider two different types of $g(\mathbf{Z}_i; \boldsymbol{\theta})$. First, $g(\mathbf{Z}_i; \boldsymbol{\theta})$ does not contain the censored observations, as given in (4.1). The second, $g(\mathbf{Z}_i; \boldsymbol{\theta})$, contains the censored observations, as given in (4.5).

In the case of uncensored observations, by Owen (1991) and Lemma 11.2 of Owen (2001), we have $\max_{1 \le i \le n} \|g(\mathbf{Z}_i; \boldsymbol{\theta}_0)\| = o_p(\sqrt{n}).$ By Lemma 1 of Tang and Leng (2012), we have under the regularity conditions R.2, R.3; the λ_{θ_0} in (2.2) satisfies $\|\lambda_{\theta_0}\| = O_p(\frac{1}{\sqrt{n}}).$ So,

$$\max_{1 \le i \le n} \left| \lambda_{\theta_0}^{\mathsf{T}} g\left(\mathbf{Z}_i; \mathbf{\theta}_0 \right) \right| = O_p \left(\frac{1}{\sqrt{n}} \right) o_p \left(\sqrt{n} \right) = o_p \left(1 \right).$$
(3.4)

Under the condition R.4; Qin and Jing (2001) proved $\max_{1 \le i \le n} \left| \lambda_{\theta_0}^{\mathsf{T}} g(\mathbf{Z}_i; \mathbf{\theta}_0) \right| = o_p(1)$ for the $g(\cdot)$ with censored observations.

Now following Owen (2001), using Taylor's series expansion of the weights, P_i 's defined in (2.2) can be rewritten as,

$$P_{i}(\boldsymbol{\theta}_{0}) = \frac{1}{n\left\{1 + \lambda_{\theta_{0}}^{\mathsf{T}}g\left(\mathbf{Z}_{i}; \boldsymbol{\theta}_{0}\right)\right\}}$$
$$= \frac{1}{n}\left[1 - \lambda_{\theta_{0}}^{\mathsf{T}}g\left(\mathbf{Z}_{i}; \boldsymbol{\theta}_{0}\right)\left\{1 + o_{p}\left(1\right)\right\}\right]$$
$$= \frac{1}{n}\left[1 - \lambda_{\theta_{0}}^{\mathsf{T}}g\left(\mathbf{Z}_{i}; \boldsymbol{\theta}_{0}\right)\right] + o_{p}\left(\frac{1}{n}\right); \quad i = 1, 2, ..., n.$$

Now we rewrite the $S_n(\boldsymbol{\beta}, \tau)$ as

$$\begin{split} S_{n}\left(\boldsymbol{\beta},\tau\right) &= \frac{1}{n}\sum_{i=1}^{n}\left[1-\lambda_{\theta_{0}}^{\mathsf{T}}g\left(\mathbf{Z}_{i};\boldsymbol{\theta}_{0}\right)\right]\mathbf{X}_{i}\left\{\mathbb{N}_{i}\left(e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(\tau)}\right)-\int_{0}^{\tau}\mathbb{I}\left[Y_{i}\geq e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(u)}\right]dH\left(u\right)\right\}+o_{p}\left(\frac{1}{n}\right)\\ &= \frac{1}{n}\sum_{i=1}^{n}\mathbf{X}_{i}\left\{\mathbb{N}_{i}\left(e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(\tau)\right)-\int_{0}^{\tau}\mathbb{I}\left[Y_{i}\geq e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(u)}\right]dH\left(u\right)\right\}\\ &-\frac{1}{n}\sum_{i=1}^{n}\lambda_{\theta_{0}}^{\mathsf{T}}g\left(\mathbf{Z}_{i};\boldsymbol{\theta}_{0}\right)\mathbf{X}_{i}\left\{\mathbb{N}_{i}\left(e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(\tau)\right)-\int_{0}^{\tau}\mathbb{I}\left[Y_{i}\geq e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(u)}\right]dH\left(u\right)\right\}+o_{p}\left(\frac{1}{n}\right)\\ &=\frac{1}{n}\sum_{i=1}^{n}\mathbf{X}_{i}\left\{\mathbb{N}_{i}\left(e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(\tau)\right)-\int_{0}^{\tau}\mathbb{I}\left[Y_{i}\geq e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(u)}\right]dH\left(u\right)\right\}\\ &-\frac{1}{n}\sum_{i=1}^{n}\mathbf{W}_{i}\left\{\mathbb{N}_{i}\left(e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(\tau)\right)-\int_{0}^{\tau}\mathbb{I}\left[Y_{i}\geq e^{\mathbf{x}_{i}^{\mathsf{T}}\boldsymbol{\beta}(u)}\right]dH\left(u\right)\right\}+o_{p}\left(\frac{1}{n}\right).\end{split}$$

Asymptotically, by (3.4) we have $\| \mathbf{W}_i \| = o_p(1); i = 1, 2, ..., n$. So,

$$S_n(\boldsymbol{\beta}, \tau) = \frac{1}{n} \sum_{i=1}^n \mathbf{X}_i \left\{ \mathbb{N}_i \left(e^{\mathbf{X}_i^{\mathsf{T}} \boldsymbol{\beta}(\tau)} \right) - \int_0^\tau \mathbb{I} \left[Y_i \ge e^{\mathbf{X}_i^{\mathsf{T}} \boldsymbol{\beta}(u)} \right] dH(u) \right\} + o_p\left(\frac{1}{n}\right).$$

Asymptotically this estimating function, $S_n(\beta, \tau)$ is equivalent to that in Peng and Huang (2008). Following the similar arguments of Peng and Huang (2008), we complete the proofs of Theorems 1 and 2.

As indicated in Peng and Huang (2008), the estimation of asymptotic variance of the quantile regression estimates is not easy since the covariance matrix of the limiting process of $\sqrt{n} \{\hat{\beta}(\tau) - \beta_0(\tau)\}$ involves unknown density function $f(y|\mathbf{X})$ and $\tilde{f}(y|\mathbf{X})$. Instead of using a smoothing or other numerical approximations, we suggest a simple bootstrap approach to estimate the standard errors of the regression estimates. This approach is used in our performance analysis discussed in next section.

4 Performance analysis

We conduct extensive simulation studies to compare the performance between our proposed EL based weighted censored quantile regression estimator and the standard censored quantile regression estimator. For our simulation, we use the models discussed in Tang and Leng (2012).

The simulation models used to generate the logarithmic event time (T_r) and logarithmic censoring time (C_r) for the r^{th} (r = 1, 2, ..., N) subject are given in Table 4.1 under four Cases (i)-(iv).

Cases	Models	Error Distribution
(i)	$T_r = \theta_0 + \theta_1 x_{1r} + \theta_2 x_{2r} + u_r,$	$u_r, v_r \sim N(0, 1)$
	$C_{r} = \gamma_{0} + \gamma_{1} x_{1r} + \gamma_{2} x_{2r} + v_{r}.$	$u_r, v_r \sim W(0, 1)$
(ii)	$T_r = \theta_0 + \theta_1 x_{1r} + \theta_2 x_{2r} + u_r,$	$H \rightarrow f(3)$
	$C_{r} = \gamma_{0} + \gamma_{1} x_{1r} + \gamma_{2} x_{2r} + v_{r}.$	$u_r, v_r \sim t(3)$
(iii)	$T_r = \theta_0 + \theta_1 x_{1r} + \theta_2 x_{2r} + (\pi_0 + \pi_0 x_{1r} + \pi_2 x_{2r}) u_r,$	$u_r, v_r \sim N(0, 1)$
	$C_{r} = \gamma_{0} + \gamma_{1}x_{1r} + \gamma_{2}x_{2r} + (\pi_{0} + \pi_{0}x_{1r} + \pi_{2}x_{2r})v_{r}.$	$u_r, v_r \sim N(0, 1)$
(iv)	$T_r = \theta_0 + \theta_1 x_{1r} + \theta_2 x_{2r} + (\pi_0 + \pi_0 x_{1r} + \pi_2 x_{2r}) u_r,$	$H \rightarrow f(3)$
	$C_{r} = \gamma_{0} + \gamma_{1}x_{1r} + \gamma_{2}x_{2r} + (\pi_{0} + \pi_{0}x_{1r} + \pi_{2}x_{2r})v_{r}.$	$u_r, v_r \sim t(3)$

 Table 4.1

 Four simulation models to generate event and censoring times

In Cases (i) and (ii), event times and censoring times are generated from the homoscedastic models and in Cases (iii) and (iv), we considered heteroscedastic models to examine the efficiency gain of our proposed method over the standard censored quantile regression. We set the parameter values as $\theta^{T} =$ $(0, -1, 0.2), \pi^{T} = (0.3, -0.1, 0.1)$ and selected γ^{T} to maintain approximately 30% of the censoring proportion in each case. We generated explanatory variables from zero mean bivariate normal distribution with covariance,

$$\Sigma = \begin{bmatrix} 1 & \sigma_{x_1, x_2} \\ \sigma_{x_1, x_2} & 1 \end{bmatrix}.$$

We considered two different ways to compute the EL based probability weights. In numerical study -I, we compute P_i 's based on the auxiliary information related to the failure time, T_i , whereas in numerical study -II, P_i 's are computed using the observed survival time, $Y_i = \min(T_i, C_i)$. In numerical study -II, we employ the synthetic variable approach (Koul et al., 1981; Qin and Jing, 2001; Li and Wang, 2003) to compute the EL based data driven probability weights.

4.1 Numerical study -I

To compute P_i 's, first we need to have a known population parameter, $\boldsymbol{\theta}$, or its estimate. We considered a linear relation between T and $\mathbf{X} = (X_1, X_2)$ with slopes (θ_1 and θ_2) and intercept (θ_0) as the auxiliary information. We estimated $\boldsymbol{\theta}$ using the standard linear regression (least square) based on a large, finite population with size, N = 10,000. We need to generate censoring times as well to compute the event indicator, $\delta_i = \mathbb{I}(T_i \leq C_i)$ and survival time, $Y_i = \min(T_i, C_i)$ to estimate the censored quantile regression parameters. To fit the weighted censored quantile regression model given in (1.2), we generated another *n* observations $\{y_i, \mathbf{x}_i\}_{i=1}^n$ with $n \ll N$, using the same models given in Table 4.1. We considered the sample sizes, n = 100 and 200 and three quantiles, $\tau = 0.25, 0.50, 0.75$. For our proposed method, we estimated P_i 's using the estimating function, $g(t_i, \mathbf{x}_i; \mathbf{0})$ defined based on the normal equations of the linear least squares method as,

$$g_i(\mathbf{z}_i; \mathbf{\theta}) = g(t_i, \mathbf{x}_i; \mathbf{\theta}) = \mathbf{x}_i(t_i - \mathbf{x}_i^{\mathsf{T}} \hat{\mathbf{\theta}}), \quad i = 1, 2, \dots, n.$$
(4.1)

For a given quantile, τ , the true value of the censored quantile regression parameters $\boldsymbol{\beta}_0(\tau)$ are estimated from the population of size, N = 10,000. In general, under a linear model assumption, the true value of the censored quantile regression slope parameters are the same as the $\boldsymbol{\theta}$ (i.e., $\beta_1(\tau) = \theta_1$, $\beta_2(\tau) = \theta_2$). But for the intercept, it is $\beta_0(\tau) = \theta_0 + F^{-1}(\tau)$, where F is the error distribution. We conducted 1,000 simulations and computed mean bias, standard error (SE) and 95% coverage probability (CP) of the model parameter estimates for different sample sizes using 250 bootstrap samples. We compared the performance of our proposed method (CQR-EL1) with the standard censored quantile regression (CQR) model. We present the simulation results in Tables 4.2 to 4.5 respectively for Cases (i)-(iv) with $\sigma_{x_1,x_2} = 0$.

				CQR			CQR-EL1	
	n	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	β_0	0.0042	0.0170	0.0647	0.0027	0.0180	0.0771
		β_1	0.0029	0.0035	0.0094	-0.0014	-0.0048	0.0030
		β_2	-0.0049	-0.0141	-0.0100	-0.0047	-0.0124	-0.0171
	200	β_0	0.0218	0.0298	0.0501	0.0199	0.0322	0.0635
		β_1	0.0016	0.0026	0.0057	0.0008	0.0028	0.0048
		β_{2}	-0.0020	-0.0032	-0.0078	-0.0010	0.0001	-0.0071
SE	100	β_0	0.1449	0.1404	0.2268	0.1103	0.1086	0.2110
		β_1	0.1533	0.1515	0.2141	0.1159	0.1109	0.2000
		β_2	0.1519	0.1525	0.2198	0.1149	0.1109	0.2082
	200	β_0	0.0973	0.0929	0.1292	0.0720	0.0703	0.1221
		β_1	0.1040	0.1029	0.1341	0.0746	0.0718	0.1173
		β_2	0.1041	0.1027	0.1354	0.0752	0.0717	0.1177
СР	100	β_0	93.3	93.4	95.7	95.8	96.6	97.0
		β_1	94.7	95.8	96.5	95.1	96.2	97.9
		β_2	96.0	96.3	96.4	96.4	96.4	96.9
	200	β_0	92.3	91.9	92.7	92.7	92.5	94.8
		β_1	94.5	96.2	95.0	95.0	95.5	96.9
		β_2	93.6	95.0	95.2	94.2	94.9	95.8

Bias, SE and CP of regression parameters for Case (i) model with independent covariates $(\sigma_{x_{i},x_{i}}=0)$

Table 4.2

			CQR			CQR-EL1		
	п	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	β_0	0.0105	0.0288	0.1088	0.0119	0.0270	0.1062
		β_1	0.0063	0.0214	0.0169	0.0005	0.0102	0.0066
		β_{2}	0.0164	0.0096	-0.0170	0.0152	0.0079	-0.0184
	200	β_0	0.0267	0.0355	0.0821	0.0276	0.0340	0.0760
		β_1	0.0006	-0.0032	0.0050	0.0042	0.0032	0.0024
		β_{2}	0.0112	0.0025	0.0051	0.0029	-0.0038	-0.0057
SE	100	β_0	0.1871	0.1538	0.2980	0.1522	0.1304	0.2914
		β_1	0.1946	0.1664	0.2698	0.1555	0.1318	0.2480
		β_{2}	0.1955	0.1676	0.2733	0.1556	0.1327	0.2543
	200	β_0	0.1235	0.1029	0.1621	0.0998	0.0871	0.1556
		β_1	0.1301	0.1146	0.1663	0.1010	0.0893	0.1473
		β_2	0.1315	0.1149	0.1671	0.1023	0.0897	0.1465
СР	100	β_0	95.5	93.1	94.7	96.2	94.8	97.2
		β_1	95.6	93.5	96.4	95.7	95.6	97.8
		β_{2}	95.9	95.4	96.4	96.0	95.0	97.2
	200	β_0	93.1	91.2	94.0	93.0	93.8	95.7
		β_1	95.0	95.5	95.4	94.8	95.5	96.2
		β_2	95.5	95.7	95.5	95.0	95.2	96.3

Table 4.3 Bias, SE and CP of regression parameters for Case (ii) model with independent covariates ($\sigma_{x_1,x_2} = 0$)

Table 4.4 Bias, SE and CP of regression parameters for Case (iii) model with independent covariates $(\sigma_{x_1, x_2} = 0)$

				CQR		CQR-EL1		
	n	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	β_0	0.0062	0.0088	0.0224	0.0055	0.0085	0.0254
		$oldsymbol{eta}_1$	0.0042	0.0051	0.0076	0.0034	0.0016	0.0057
		β_2	-0.0038	-0.0039	-0.0069	-0.0013	0.0003	-0.0010
	200	${m eta}_0$	0.0064	0.0072	0.0167	0.0064	0.0089	0.0195
		β_1	0.0012	0.0038	0.0033	-0.0006	-0.0003	-0.0014
		β_2	-0.0015	-0.0031	-0.0017	-0.0004	0.0002	0.0023
SE	100	β_0	0.0472	0.0466	0.0767	0.0349	0.0338	0.0737
		β_1	0.0566	0.0570	0.0796	0.0424	0.0411	0.0708
		β_2	0.0567	0.0575	0.0807	0.0425	0.0418	0.0720
	200	β_0	0.0313	0.0301	0.0402	0.0225	0.0213	0.0345
		β_1	0.0371	0.0377	0.0489	0.0276	0.0267	0.0402
		β_2	0.0367	0.0376	0.0488	0.0270	0.0267	0.0401
СР	100	${m eta}_0$	94.4	95.0	96.1	94.3	96.0	97.1
		$oldsymbol{eta}_1$	95.0	95.2	95.5	95.2	95.3	97.4
		β_2	96.6	96.7	97.3	95.4	96.6	98.0
	200	${oldsymbol{eta}}_0$	94.1	93.4	94.9	93.2	94.0	94.1
		β_1	94.0	94.9	96.0	93.0	95.1	95.9
		β_{2}	94.6	95.0	95.3	94.4	95.3	94.8

95.4

95.2

95.2

96.0

				CQR			CQR-EL1	
	n	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	${m eta}_0$	0.0066	0.0097	0.0364	0.0048	0.0076	0.0273
		β_1	0.0031	0.0039	0.0041	0.0026	0.0043	0.0036
		β_2	0.0008	-0.0009	-0.0018	0.0008	-0.0035	-0.0028
	200	${m eta}_0$	0.0083	0.0089	0.0243	0.0100	0.0103	0.0258
		β_1	-0.0020	0.0016	0.0017	-0.0022	-0.0008	-0.0018
		β_2	0.0008	-0.0012	-0.0031	0.0026	0.0012	0.0004
SE	100	β_0	0.0600	0.0507	0.1103	0.0466	0.0407	0.1038
		β_1	0.0667	0.0592	0.0993	0.0514	0.0468	0.0885
		β_2	0.0677	0.0600	0.1014	0.0525	0.0470	0.0921
	200	${m eta}_0$	0.0395	0.0327	0.0521	0.0305	0.0260	0.0464
		β_1	0.0429	0.0386	0.0568	0.0331	0.0298	0.0491
		β_2	0.0429	0.0389	0.0580	0.0331	0.0301	0.0501
СР	100	${m eta}_0$	93.5	95.0	97.7	94.7	95.5	97.8
		β_1	95.6	96.6	97.0	96.0	96.3	97.3
		β_2	96.0	96.2	97.3	95.8	96.7	97.0
	200	β_0	93.0	93.9	94.9	93.5	93.4	94.1

95.8

95.9

94.7

95.5

94.5

94.5

 β_1

ß,

95.6

94.5

Table 4.5 Bias, SE and CP of regression parameters for Case (iv) model with independent covariates $(\sigma_{y}) = 0$

From Tables 4.2-4.5, we see that our proposed estimator has approximately zero bias. A comparison of SE of CQR-EL1 with CQR indicates that the SE of CQR-EL1 reduces remarkably for all the parameters irrespective of any quantile. For example, we consider the scenario of n = 100 and $\tau = 0.25$ for comparison purposes throughout this paper. From Table 4.2, for CQR, SE of $\hat{\beta}_1$ is 0.1533 and for CQR-EL1, SE of $\hat{\beta}_1$ is reduced to 0.1159. When the sample size is increased to 200, SE of $\hat{\beta}_1$ of our proposed method further is reduced to 0.0746. If we compare the CP of our proposed method with the nominal level of 95%, CQR-EL1 provides approximately 95% coverage and becomes more stable when the sample size increases. Similar conclusions can be reached for Case (ii) (results are in Table 4.3) even though we considered heavy tailed distribution for the failure time compared to Case (i). For example, SE of $\hat{\beta}_1$ using CQR is 0.1946, whereas it is only 0.1555 for the CQR-EL1 based estimate. We also observed that SE is comparatively high in Case (ii) compared to Case (i).

In Cases (iii) and (iv), the error depends on the covariates. Simulation results for these Cases (Tables 4.4 and 4.5) are almost similar to the cases where error is independent of covariates. For example, in Case (iii) (Table 4.4), SE of $\hat{\beta}_1$ is 0.0566 and 0.0424 for CQR and CQR-EL1 respectively. Similarly, in Case (iv) (Table 4.5), SE of $\hat{\beta}_1$ is 0.0667 and 0.0514 for CQR and CQR-EL1 respectively. Here, we could also see a slight increase in the SE of estimates for Case (iv) because of the heavy tailed distribution assumption for the failure time compared to Case (iii).

4.2 Numerical study -II

In most of the survival data with random right censoring, the observed data are the triplet $\{Y = \min(T, C), \mathbf{X}, \delta\}$. We consider a linear relationship between the survival time (Y) and the covariates as the auxiliary information. Here we cannot use the EL estimating function, $g(\cdot)$ defined in (4.1) because of the censoring. There are other methods available in the literature which take care of the right censoring in the linear regression.

Koul et al. (1981) introduced a synthetic data approach by transforming the survival time, Y_r to a synthetic variable, \tilde{Y}_r as

$$\tilde{Y}_{r} = \frac{\delta_{r} Y_{r}}{1 - G(Y_{r})}; \quad r = 1, 2, ..., N,$$
(4.2)

where δ_r is the censoring indicator and $G(\cdot)$ is the distribution of the censoring time. $E(\tilde{Y} | \mathbf{X}) = E(Y | \mathbf{X})$ if *C* is independent of both **X** and *Y*. When $G(\cdot)$ is unknown, we can replace it with its Kaplan-Meier estimator. The estimator of $G(\cdot)$ using the Kaplan-Meier (Kaplan and Meier, 1958) estimator is

$$1 - \hat{G}_{N}(t) = \prod_{r=1}^{N} \left(\frac{N-r}{N-r+1} \right)^{\mathbb{I}(Y_{(r)} \le t, \ \delta_{(r)} = 0)},$$
(4.3)

where $Y_{(r)}$'s are ordered and the corresponding censoring indicator is $\delta_{(r)}$. We can estimate θ as

$$\tilde{\boldsymbol{\theta}} = \left(\mathbf{X}^{\mathsf{T}}\mathbf{X}\right)^{-1}\mathbf{X}^{\mathsf{T}}\tilde{Y}_{r}.$$
(4.4)

Qin and Jing (2001) and Li and Wang (2003) independently provided the estimating function to compute the EL based data driven probabilities as

$$g_i\left(\mathbf{z}_i; \tilde{\mathbf{\theta}}\right) = g\left(y_i, \mathbf{x}_i, \delta_i; \tilde{\mathbf{\theta}}\right) = \mathbf{x}_i\left(\tilde{y}_i - \mathbf{x}_i^{\mathsf{T}}\tilde{\mathbf{\theta}}\right), \quad i = 1, 2, \dots, n.$$
(4.5)

We can compute the \tilde{y}_i and $\hat{G}_n(t)$ using the sample analogues of (4.2) and (4.3) respectively.

To compute P_i 's, we consider a linear relation between Y and $\mathbf{X} = (X_1, X_2)$ with slopes $(\theta_1 \text{ and } \theta_2)$ and intercept (θ_0) . We estimate $\mathbf{\theta}$ using (4.4) based on a large, finite population with size, N = 10,000. To fit the weighted censored quantile regression model given in (1.2), we generate another *n* observations $\{y_i, \mathbf{x}_i\}_{i=1}^n$ with $n \ll N$ using the same models given in Table 4.1. For our proposed method, we estimate P_i 's using the estimating function, $g(y_i, \mathbf{x}_i, \delta_i; \tilde{\mathbf{\theta}})$ given in (4.5).

Similar to numerical study -I, we present the results based on 1,000 simulations and report the bias, standard error (SE) and empirical coverage probability (CP) for the nominal level of 95% based on 250 bootstrap samples. We provide the summary of the simulation results for this study in Tables 4.6-4.9.

Similar to the population information related to T (numerical study -I), conclusions are almost similar for uncorrelated covariates. From Tables 4.6-4.9 we see that our proposed method (CQR-EL2) provides unbiased estimates irrespective of any sample size and quantile. If we consider the coverage probability,

both CQR and CQR-EL2 provide approximately 95% coverage. For any quantile, there is a reduction in the standard error of CQR-EL2 parameter estimates compared to CQR parameter estimates. If we consider Case (i) as a basic model, CQR-EL2 with Case (ii) has reasonably higher SE along with CQR because of the heavy tailed distribution of the observed survival time. When the error depended on the covariates (Cases (iii) & (iv)), the SE of CQR-EL2 reduced considerably.

We also conducted large number of simulations with correlated covariates with $\sigma_{x_1, x_2} = 0.5$ as well as constructed weights based on simple relationship with one covariate only for both numerical studies. The results of these simulations are not provided here to save the space. The conclusions arrived are almost similar to the uncorrelated covariate cases.

In numerical study -I, we noticed that there is a slight reduction in SE of $\hat{\beta}_2$ using heteroscedastic models for CQR-EL1. But use of the estimating function, $g(y_i, x_{1i}, \delta_i; \tilde{\theta})$ (CQR-EL2), does not reduce the SE of $\hat{\beta}_2$ under heteroscedastic models. Since we utilized only partial population information in relation to X_1 , the standard error of $\hat{\beta}_0$ and $\hat{\beta}_1$ reduced for CQR-EL2 compared to CQR. The standard error of $\hat{\beta}_2$ was not changed.

Our simulation studies reveal that auxiliary information greatly enhances the efficiency of estimation, if the population information related to both X_1 and X_2 is available. If the population information is only related to X_1 , the efficiency gain is limited to β_0 and β_1 only. However, under heteroscedastic models, the efficiency of estimating β_2 slightly improved in numerical study -I, but not in numerical study -II.

,	8	-			1		(x_1, x_2)	-)
				CQR			CQR-EL2	
	n	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	${\boldsymbol eta}_0$	0.0042	0.0170	0.0647	0.0217	0.0275	0.0720
		β_1	0.0029	0.0035	0.0094	-0.0491	-0.0411	-0.0090
		β_{2}	-0.0049	-0.0141	-0.0100	0.0116	-0.0029	-0.0194
	200	${\boldsymbol eta}_0$	0.0218	0.0298	0.0501	0.0220	0.0323	0.0562
		β_1	0.0016	0.0026	0.0057	-0.0295	-0.0273	-0.0119
		β_{2}	-0.0020	-0.0032	-0.0078	0.0034	0.0053	-0.0011
SE	100	${\boldsymbol eta}_0$	0.1449	0.1404	0.2268	0.1273	0.1233	0.2160
		β_1	0.1533	0.1515	0.2141	0.1475	0.1416	0.2075
		β_2	0.1519	0.1525	0.2198	0.1416	0.1414	0.2162
	200	${\boldsymbol eta}_0$	0.0973	0.0929	0.1292	0.0840	0.0798	0.1239
		β_1	0.1040	0.1029	0.1341	0.0970	0.0921	0.1278
		β_{2}	0.1041	0.1027	0.1354	0.0957	0.0936	0.1304
СР	100	${\boldsymbol eta}_0$	93.3	93.4	95.7	94.3	96.1	96.8
		β_1	94.7	95.8	96.5	94.6	96.1	96.9
		β_2	96.0	96.3	96.4	95.4	95.4	97.4
	200	${m eta}_0$	92.3	91.9	92.7	92.9	92.3	94.3
		β_1	94.5	96.2	95.0	95.3	95.3	94.8
		β_2	93.6	95.0	95.2	93.5	94.9	95.9

Bias, SE and CP of regression parameters for Case (i) model with independent covariates ($\sigma_{x_1,x_2} = 0$)

Table 4.6

				CQR			CQR-EL2	
	n	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	β_0	0.0105	0.0288	0.1088	0.0306	0.0461	0.1139
		β_1	0.0063	0.0214	0.0169	-0.0841	-0.0503	-0.0216
		β_{2}	0.0164	0.0096	-0.0170	0.0329	0.0260	-0.0094
	200	β_0	0.0267	0.0355	0.0821	0.0419	0.0508	0.0921
		β_1	0.0006	-0.0032	0.0050	-0.0022	-0.0010	-0.0188
		β_2	0.0112	0.0025	0.0051	0.0251	0.0137	0.0133
SE	100	β_0	0.1871	0.1538	0.2980	0.1619	0.1379	0.2768
		β_1	0.1946	0.1664	0.2698	0.1863	0.1595	0.2548
		β_2	0.1955	0.1676	0.2733	0.1787	0.1549	0.2632
	200	β_0	0.1235	0.1029	0.1621	0.1048	0.0900	0.1551
		β_1	0.1301	0.1146	0.1663	0.1214	0.1052	0.1575
		β_2	0.1315	0.1149	0.1671	0.1185	0.1044	0.1606
СР	100	β_0	95.5	93.1	94.7	95.9	94.2	97.5
		β_1	95.6	93.5	96.4	94.8	93.3	96.7
		β_2	95.9	95.4	96.4	94.2	94.2	96.3
	200	β_0	93.1	91.2	94.0	93.5	93.0	94.7
		β_1	95.0	95.5	95.4	94.5	94.0	94.9
		β_2	95.5	95.7	95.5	94.8	94.5	95.4

Table 4.7 Bias, SE and CP of regression parameters for Case (ii) model with independent covariates ($\sigma_{x_1, x_2} = 0$)

Table 4.8 Bias, SE and CP of regression parameters for Case (iii) model with independent covariates $(\sigma_{x_1, x_2} = 0)$

								-
				CQR			CQR-EL2	
	п	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	β_0	0.0062	0.0088	0.0224	0.0127	0.0146	0.0302
		β_1	0.0042	0.0051	0.0076	-0.0071	-0.0043	0.0021
		β_2	-0.0038	-0.0039	-0.0069	0.0018	0.0017	-0.0040
	200	β_0	0.0064	0.0072	0.0167	0.0094	0.0105	0.0197
		β_1	0.0012	0.0038	0.0033	-0.0042	-0.0026	-0.0007
		β_2	-0.0015	-0.0031	-0.0017	0.0009	-0.0003	0.0015
SE	100	β_0	0.0472	0.0466	0.0767	0.0448	0.0445	0.0801
		β_1	0.0566	0.0570	0.0796	0.0541	0.0549	0.0830
		β_2	0.0567	0.0575	0.0807	0.0538	0.0558	0.0833
	200	β_0	0.0313	0.0301	0.0402	0.0292	0.0283	0.0396
		β_1	0.0371	0.0377	0.0489	0.0348	0.0356	0.0484
		β_2	0.0367	0.0376	0.0488	0.0344	0.0359	0.0488
СР	100	β_0	94.4	95.0	96.1	93.9	94.7	96.9
		β_1	95.0	95.2	95.5	94.6	94.7	96.3
		β_2	96.6	96.7	97.3	95.8	96.4	97.3
	200	β_0	94.1	93.4	94.9	93.9	93.8	94.9
		β_1	94.0	94.9	96.0	94.1	94.3	95.0
		β_2	94.6	95.0	95.3	94.0	95.4	94.3

				CQR			CQR-EL2	
	n	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
Bias	100	β_0	0.0066	0.0097	0.0364	0.0189	0.0169	0.0419
		β_1	0.0031	0.0039	0.0041	-0.0138	-0.0073	-0.0000
		β_{2}	0.0008	-0.0009	-0.0018	0.0074	0.0060	0.0024
	200	β_0	0.0083	0.0089	0.0243	0.0124	0.0119	0.0273
		β_1	-0.0020	0.0016	0.0017	-0.0097	-0.0051	-0.0032
		β_{2}	0.0008	-0.0012	-0.0031	0.0019	0.0004	-0.0020
SE	100	β_0	0.0600	0.0507	0.1103	0.0548	0.0486	0.1159
		β_1	0.0667	0.0592	0.0993	0.0618	0.0581	0.1018
		β_{2}	0.0677	0.0600	0.1014	0.0616	0.0578	0.1066
	200	β_0	0.0395	0.0327	0.0521	0.0359	0.0304	0.0516
		β_1	0.0429	0.0386	0.0568	0.0397	0.0364	0.0558
		β_{2}	0.0429	0.0389	0.0580	0.0397	0.0368	0.0579
СР	100	β_0	93.5	95.0	97.7	92.9	95.2	97.6
		β_{1}	95.6	96.6	97.0	94.2	95.5	97.4
		β_{2}	96.0	96.2	97.3	96.3	97.0	97.6
	200	β_0	93.0	93.9	94.9	93.3	94.2	95.8
		β_1	95.6	95.8	94.7	94.0	95.5	95.2
		β_2	94.5	95.9	95.5	94.9	96.0	94.7

Table 4.9 Bias, SE and CP of regression parameters for Case (iv) model with independent covariates $(\sigma_{x_1, y_2} = 0)$

Note that the value of the auxiliary parameter value plays a big role in the efficiency of the weighted censored quantile regression parameter estimates. If the estimate of θ based present study data and previous study (or known θ value) are very close, then all weights will be close to 1/n and solutions to (1.1) and (1.2) remain the same. If data on previous studies are not available, we can make of the data available in the present study to estimate the value of θ . In this case, if dimensions of θ and estimating equation $g(z, \theta)$ are same, then all weights will be equal to 1/n and solutions to (1.1) and (1.2) remain same. However, if the dimensions of $g(z, \theta)$ is greater than that of θ , the weights $p(\hat{\theta})$ is no longer equal to 1/n and this scheme provides an efficiency gain over the conventional QR estimates (Tang and Leng, 2012).

4.3 Case example

The North Central Cancer Treatment Group (NCCTG) was initiated by a group of physicians from the north central region of the United States of America and the Mayo Clinic in Rochester, Minnesota. This study was conducted by NCCTG to determine whether the conclusions from the patient-completed questionnaire and those already obtained by the patient's physician were independent or not (Loprinzi, Laurie, Wieand, Krook, Novotny, Kugler, Bartel, Law, Bateman and Klatt, 1994). They used the performance scores (ECOG and Karnofsky) to assess the patient's daily activities. The dataset is available in the "survival" package of R software with readings of 228 patients. Because of the incompleteness of some of the variables, we had to limit the dataset to 167 observations. For the illustration of our proposed method, we changed our focus to identify the effect of following covariates over the observed survival time at different quantiles. We considered "age", patient's age in years; "sex", (Male = 1 Female = 2); "ph.ecog",

ECOG performance score measured by physician (0 = good 5 = dead); "meal.cal", calories consumed at meals and "wt.loss", weight loss in the last six months as the covariates. After removing the incomplete patient readings, the available ECOG scores were 0,1 and 2 only. We defined two dummy categorical variables for "ph.ecog" as follows.

$$ecog1 = \begin{cases} 1, & \text{if ph.ecog} = 1 \\ 0, & \text{otherwise} \end{cases}$$
$$ecog2 = \begin{cases} 1, & \text{if ph.ecog} = 2 \\ 0, & \text{otherwise.} \end{cases}$$

To demonstrate the usefulness of our proposed method, we randomly selected a part (100 observations) of the complete data (167 observations) by considering it to be the data available from the previous study. We assumed that there exists a linear relation between the logarithm of the observed survival time and all the continuous explanatory variables (age, meal.cal and wt.loss) as the available auxiliary information. We estimated the $\mathbf{\theta} = (\theta_0, \theta_{age}, \theta_{meal}, \theta_{wt})$ by the least square method based on 100 observations where the response is the synthetic variable defined by (4.2). Then we computed the EL based data driven probability weights for the present study data points (67 observations). After computing the weights, we estimated the weighted censored quantile regression parameters using Peng and Huang (2008) method with all the covariates. For the present study data, the censoring proportion is 0.283. Interestingly, we estimated the regression parameters using CQR up to the 86th quantile, where as we could estimate to the 90th quantile using CQR-EL2. Along with the estimates for the quantiles, $\tau = 0.25$, 0.50, 0.75, we report standard error (SE) and 95% confidence limits using 250 bootstrap samples as well in Table 4.10.

			CQR			CQR-EL2	
	$\tau \rightarrow$	0.25	0.50	0.75	0.25	0.50	0.75
β	Intercept	5.4777	4.2651	5.5380	4.7531	4.1729	6.4258
P	Age	-0.0168	0.0179	0.0040	-0.0047	0.0202	-0.0032
	Sex	0.7201	0.6180	0.4181	0.7606	0.6638	0.3651
	ECOG1	-0.7059	-0.5449	-0.2029	-0.5701	-0.5355	-0.2884
	ECOG2	-0.8677	-0.9402	-0.8336	-1.1584	-1.0612	-1.0192
	MealCal	0.0004	0.0001	0.0001	0.0004	0.0001	-0.0000
	WtLoss	-0.0007	-0.0084	-0.0023	-0.0023	-0.0100	-0.0135
SE	Intercept	1.9235	1.4314	1.7494	1.6628	1.4149	1.4666
	Age	0.0277	0.0188	0.0225	0.0256	0.0184	0.0176
	Sex	0.5610	0.3389	0.3716	0.5374	0.3317	0.2809
	ECOG1	0.6521	0.3436	0.3375	0.6498	0.3493	0.2434
	ECOG2	1.0317	0.5410	0.6061	0.9336	0.5413	0.3879
	MealCal	0.0009	0.0006	0.0008	0.0009	0.0006	0.0005
	WtLoss	0.0181	0.0128	0.0231	0.0157	0.0124	0.0100
CI	Intercept	(1.6, 9.14)	(2.38, 8)	(2.08, 8.94)	(1.79, 8.31)	(2.32, 7.87)	(3.14, 8.89)
	Age	(-0.07, 0.04)	(-0.04, 0.04)	(-0.04, 0.05)	(-0.06, 0.04)	(-0.03, 0.04)	(-0.03, 0.04)
	Sex	(-0.45, 1.74)	(0, 1.33)	(-0.13, 1.33)	(-0.39, 1.71)	(-0.04, 1.27)	(-0.07, 1.03)
	ECOG1	(-1.75, 0.81)	(-1.15, 0.2)	(-0.97, 0.35)	(-1.86, 0.69)	(-1.18, 0.19)	(-0.78, 0.18)
	ECOG2	(-2.88, 1.16)	(-2, 0.12)	(-2.11, 0.26)	(-2.83, 0.83)	(-2.13, -0.01)	(-1.73, -0.21)
	WtLoss	(-0.04, 0.03)	(-0.03, 0.02)	(-0.05, 0.04)	(-0.04, 0.02)	(-0.03, 0.01)	(-0.04, 0)

 Table 4.10

 Estimates, SE and 95% CI for regression parameters of NCCTG lung cancer data

From Table 4.10, we see that the standard error of the estimates of all the continuous variable parameters and the intercept reduced considerably because we considered the auxiliary information related to them. For the remaining variables, a reduction of standard error can also be seen, even though we did not consider any auxiliary information related to them. In the censored quantile regression with the EL based data driven probability weights, we see narrower 95% confidence limits for all the variables compared to those using the standard censored quantile regression.

5 Conclusions

We proposed a method which effectively use the auxiliary information to improve the efficiency of the censored quantile regression estimator. We developed a methodology to transform the population information available from previous clinical trials or from some existing facts into non-parametric empirical likelihood based data driven probabilities. We developed the EL based data driven probability computation for both known and unknown cases of prior information regarding population parameters. We applied these probabilities as the weights into Peng and Huang (2008) censored quantile regression model. Our proposed method is efficient compared to standard censored quantile regression and provides consistent estimators of regression coefficients with asymptotic normality. Our simulations studies showed that the standard error of the parameter estimates based on our proposed methods (CQR-EL1 and CQR-EL2) is lower than the standard method (CQR) when we use all the covariates for computing the EL based data driven probability weights. Our proposed weighted censored quantile regression method provides almost the same coverage probability compared to the nominal level. In the case of heteroscedastic models, even the use of the auxiliary information regarding a subset of population parameters improved the efficiency of the estimates of all the parameters by using CQR-EL1. But in CQR-EL2, the efficiency improvement was limited to the corresponding subset of variables and intercept. In homoscedastic models, the use of auxiliary information regarding a subset of population parameters improved the efficiency only for that particular subset of parameters and the intercept in both CQR-EL1 and CQR-EL2. In the real data analysis, we observed that our proposed method provides more efficient quantile estimates and narrower confidence limits compared to the standard censored quantile regression.

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Abstract

Item nonresponse is frequently encountered in sample surveys. Hot-deck imputation is commonly used to fill in missing item values within homogeneous groups called imputation classes. We propose a fractional hot-deck imputation procedure and an associated empirical likelihood for inference on the population mean of a function of a variable of interest with missing data under probability proportional to size sampling with negligible sampling fractions. We derive the limiting distributions of the maximum empirical likelihood estimator and empirical likelihood ratio, and propose two related asymptotically valid bootstrap procedures to construct confidence intervals for the population mean. Simulation studies show that the proposed bootstrap procedures outperform the customary bootstrap procedures which are shown to be asymptotically incorrect when the number of random draws in the fractional imputation is fixed. Moreover, the proposed bootstrap procedure based on the empirical likelihood ratio is seen to perform significantly better than the method based on the limiting distribution of the maximum empirical likelihood estimator when the inclusion probabilities vary considerably or when the sample size is not large.

Key Words: Asymptotic distribution; Bootstrap; Confidence interval; Empirical likelihood ratio; Imputation; PPS sampling.

1 Introduction

Item nonresponse is commonly seen in sample surveys. A popular method of handling item nonresponse is hot-deck imputation because (i) it preserves the distribution of item values as opposed to mean imputation which leads to a "spike" at the mean of respondent values, (ii) it provides a complete data file and allows the same survey weight to be used for all items, and (iii) results from different analyses based on imputed data are consistent with each other (Rao and Shao, 1992).

Our focus is on fractional hot-deck imputation, where a few values are drawn randomly from the set of respondent values (donors) and the average or weighted average of the drawn values is used to fill in a missing value. For validity and accuracy of inference based on imputation, the observed sample is usually grouped into homogeneous classes, called imputation classes, according to auxiliary variables that are observed for all the sample units (Brick and Kalton, 1996). Haziza and Beaumont (2007) gave a comprehensive review on different methods of constructing imputation classes. Missing values are imputed using the donors within classes and independently across classes.

We consider the case where imputation classes are formed according to a categorical variable z with finite support $\{1, ..., K\}$ and whose value is observed on all the units of a probability sample, denoted s, of fixed size n, selected from a finite population U of size N. We shall focus on the probability proportional to size (PPS) sampling with replacement or without replacement with negligible sampling fractions in the paper, but our theory applies to any fixed-size unequal probability sampling design with

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replacement. Let y be a variable (or item) of interest with value y_i observed only for some but not all $i \in s$ due to nonresponse. Let δ_i be the response indicator for y_i taking the value 1 when the associated y_i is observed and 0 otherwise. The δ_i are assumed to be independent random variables across $i \in s$. We assume that y_i is missing at random (MAR) with uniform response rate within each imputation class, that is, given the value of z_i , the probability of response of unit *i* does not depend on the value of y_i . More precisely, this means that, for k = 1, ..., K,

$$P(\delta_i = 1 | y_i, z_i = k) = P(\delta_i = 1 | z_i = k) = P_k$$

for all $i \in s$ such that $z_i = k$, where $P_k \in (0, 1)$ is a constant.

We aim to construct reliable confidence intervals (CIs) for the population mean μ of a given function h of y, i.e.,

$$\mu = \frac{1}{N} \sum_{i \in U} h(y_i).$$

For example, taking h(y) = y yields the population mean of y, and taking $h(y) = 1(y \le t)$, where $1(\cdot)$ is the indicator function, gives the finite-population distribution function of y at a given value t. Note that μ can be alternatively defined as the solution to the population-level estimating equation $\sum_{i \in U} (h(y_i) - \mu) = 0$ in μ , and similarly, our theory can be readily extended to constructing CIs for a population parameter θ defined by the solution to the equation $\sum_{i \in U} g(y_i, \theta) = 0$ for a general smooth estimating function $g(y, \theta)$ in θ .

The empirical likelihood (EL) method, proposed by Owen (1988) for independent and identically distributed (IID) complete data, has received much attention since it provides a non-parametric approach to constructing likelihood-ratio-type confidence intervals. The EL ratio intervals have several desirable properties: shape and orientation of these intervals are determined entirely by the data and the intervals are range preserving and transformation invariant (Owen, 2001). Qin and Lawless (1994) studied EL inference for parameters defined by smooth estimating equations. Wang and Chen (2009) used EL and imputation to handle IID data that are subject to a MAR assumption. Tang and Qin (2012) proposed an efficient EL estimator based on an inverse-probability-weighted imputation for IID MAR data. In the sample survey context, several variants of EL have been proposed. Chen and Sitter (1999) developed a pseudo EL for complex surveys with auxiliary information. Chen and Kim (2014) proposed a population EL. However, neither papers handles the case of missing data. Cai, Qin, Rao and Winiszewska (2019) proposed an EL method based on imputation for missing survey data under stratified random sampling. Our paper adapts the method of Cai et al. (2019) to accommodate unequal probability sampling designs.

We define a fractionally imputed estimating function of the mean parameter μ and propose an EL method based on this imputed estimating function for PPS sampling with negligible sampling fractions or with replacement. We derive the asymptotic distributions of the associated maximum EL estimator (MELE) and EL ratio. Based on these limiting distributions, we propose two asymptotically correct bootstrap

methods for constructing CIs on μ . Additionally, we show that the usual bootstrap procedures lead to asymptotically incorrect coverage probabilities when the number of random draws in the fractional imputation is fixed. Simulation studies show that the proposed EL-ratio-based bootstrap interval clearly outperforms the proposed MELE-based bootstrap interval especially when the inclusion probabilities vary considerably or when the sample size is not too large.

Section 2 introduces the proposed fractional imputation on a function of the population mean. Section 3 presents an EL under imputation and its asymptotic properties. Section 4 gives the proposed bootstrap-EL procedures for constructing CIs. Section 5 presents results from simulation studies. Lengthy technical details and proofs are delegated to the Appendices.

2 Fractional imputation

Following the notation in Section 1, for inference about the population mean μ of h(y), we first impute the missing y_i within imputation classes as follows. Let $p_i, i \in U$, be the probabilities induced by a positive size measure, according to which a PPS sample *s* is selected. Define $d_i = 1/(np_i)$ for all $i \in U$. We use d_i as design weights in the paper. Define the donor set \mathcal{R}_k of class k, k = 1, ..., K, as the set of pairs (y_i, d_i) within class *k* for which y_i are observed, that is,

$$\mathcal{R}_{k} = \{(y_{i}, d_{i}): i \in s, \delta_{i} = 1, z_{i} = k\}.$$

For an $i \in s$ with $z_i = k$, if $\delta_i = 0$, we select $J \ge 1$ pairs of (y_i, d_i) at random with replacement from \mathcal{R}_k and denote them as (y_{ij}^*, d_{ij}^*) for j = 1, ..., J. We then define an *imputed estimating function* of μ for all $i \in s$ as

$$\tilde{h}_{i}(\mu) = \sum_{k=1}^{K} \mathbb{1}(z_{i} = k) \bigg[\delta_{i} d_{i} \{h(y_{i}) - \mu\} + (1 - \delta_{i}) J^{-1} \sum_{j=1}^{J} d_{ij}^{*} \{h(y_{ij}^{*}) - \mu\} \bigg], \quad (2.1)$$

where $1(z_i = k)$ is an indicator function taking value 1 if $z_i = k$ and 0 otherwise. This is essentially fractional imputation (Kalton and Kish, 1984) on the function $d_i \{h(y_i) - \mu\}$. Note that the typical approach to imputing a missing value from an unequal probability sample, as proposed by Rao and Shao (1992), is to draw y – values from the donor set within imputation class k with probabilities $d_i / \sum_{i \in \mathcal{R}_k} d_i$, then use the drawn values as the imputed values for the missing observation. As oppose to that approach, we draw y values from the donor set with equal probabilities and attach the corresponding design weights as factors to obtain imputed values. Our approach agrees with that of Platek and Gray (1983).

The number of random draws J in (2.1) is assumed to be a fixed integer that does not change with sample size n. This is a typical setup that is used in most real-world applications. In fact, single random imputation with J = 1 is often used in practice. This setting distinguishes our study from most studies in the literature (such as Wang and Chen (2009)), where J is assumed to increase with n to infinity in asymptotic studies. Note that, in the survey context, imputed values are reported along with the observed ones in data files. Having a small J will keep the data file manageable, which is the primary reason that J = 1 is preferred in practice. Having a fixed J also frees users from choosing appropriate J for asymptotic validity. In addition, using a small J lessens computation time, although not substantially with modern computers unless both the sample size and the proportion of missing are very large.

3 EL inference under imputation

Based on the imputed function $\tilde{h}_i(\mu)$ for all $i \in s$, we now propose an EL method for inference about the population mean μ of $h(y_i)$. Define the EL under imputation (2.1) as $L_n(q) = \prod_{i \in s} q_i$, where q_i satisfy $q_i \ge 0$ for all i, $\sum_{i \in s} q_i = 1$, and $\sum_{i \in s} q_i \tilde{h}_i(\mu) = 0$. The corresponding profile log-EL at a given value of μ is defined as

$$l_{n}(\mu) = \sup_{\{q_{i}\}} \left\{ \sum_{i \in s} \log q_{i} : q_{i} \ge 0, \sum_{i \in s} q_{i} = 1, \sum_{i \in s} q_{i} \tilde{h}_{i}(\mu) = 0 \right\}.$$
(3.1)

Note that although the above $l_n(\mu)$ takes the same form as the EL of the mean for IID data (Owen, 2001), the design weights d_i are in fact subsumed in the definition (2.1) of $\tilde{h}_i(\mu)$, which makes $l_n(\mu)$ suitable for survey data. Solving the above maximization problem using the method of Lagrange multipliers, we obtain

$$q_{i} = \frac{1}{n\left\{1 + \lambda \tilde{h}_{i}(\mu)\right\}}$$

where λ is the solution to the equation $n^{-1}\sum_{i\in s}\tilde{h}_i(\mu)/\{1+\lambda\tilde{h}_i(\mu)\}=0$. Consequently, we get

$$l_n(\mu) = -\sum_{i \in s} \log \left\{ 1 + \lambda \tilde{h}_i(\mu) \right\}.$$
(3.2)

We then define the MELE of μ as

$$\hat{\mu} = \operatorname{argmax}_{\mu} l_n(\mu).$$

It can be shown that the maximum of $l_n(\mu)$ is attained when $q_i = 1/n$ for all $i \in s$. Hence, by the third constraint in (3.1), the MELE $\hat{\mu}$ is the solution to the equation $n^{-1}\sum_{i \in s} \tilde{h}_i(\mu) = 0$, which is given by

$$\hat{\mu} = \frac{\sum_{i \in s} \sum_{k=1}^{K} \mathbb{1}(z_i = k) \left\{ \delta_i d_i h(y_i) + (1 - \delta_i) J^{-1} \sum_{j=1}^{J} d_{ij}^* h(y_{ij}^*) \right\}}{\sum_{i \in s} \sum_{k=1}^{K} \mathbb{1}(z_i = k) \left\{ \delta_i d_i + (1 - \delta_i) J^{-1} \sum_{j=1}^{J} d_{ij}^* \right\}}.$$
(3.3)

Our Theorem 1 below presents the asymptotic normality of the MELE $\hat{\mu}$. For the asymptotic investigation, we consider the case where both the population size N and the sample size n increase to ∞ as an index ν increases to ∞ , as assumed by Chen and Rao (2007).

Theorem 1. Assume that the regularity conditions (R.1)-(R.2) in Appendix A hold. Under PPS sampling with replacement,

$$\sqrt{n}\sigma_N^{-1}(\hat{\mu}-\mu_N) \stackrel{d}{\rightarrow} N(0,1)$$

as $n \to \infty$, where μ_N is the true parameter value and σ_N is a constant.

An expression of σ_N is given in Appendix A and the proof of Theorem 1 is given in Appendix B. Theorem 1 also applies to PPS sampling without replacement with negligible sampling fractions, which is asymptotically equivalent to PPS sampling with replacement. Note that increasing J reduces the asymptotic variance σ_N , leading to a more efficient estimator. However, a larger J does not necessarily imply better coverage probabilities of CIs, as shown in the simulation study in Section 5.

Theorem 1 suggests that we can construct a Wald-type CI for μ_N if given a design-consistent estimator of σ_N . However, accurately estimating σ_N under an unequal probability sampling design is not an easy task, especially when σ_N has a complicated algebraic expression. An alternative approach is to construct a likelihood-ratio-like quantity as in parametric likelihood inference. Toward this end, we define an *EL ratio* as

$$R(\mu) = -2l_n(\mu).$$
(3.4)

The asymptotic distribution of $R(\mu)$ is given by Theorem 2.

Theorem 2. Assume that the regularity conditions (R.1)-(R.3) in Appendix A hold. Under PPS sampling with replacement,

$$R(\mu_N) \stackrel{d}{\rightarrow} c_N \chi_1^2$$

as $n \to \infty$, where χ_1^2 is a chi-square random variable with 1 degree of freedom and c_N is a constant depending on μ_N .

The proof of Theorem 2 is given in Appendix B. Again, Theorem 2 also applies to PPS sampling without replacement with negligible sampling fractions. The expression of the scaling constant c_N is given in Appendix A. The value of c_N is not 1 in general. Hence, to construct an EL ratio CI for μ_N , we need to estimate an unknown constant as in the case of Wald-type test. A design-consistent estimator of c_N is given in Appendix A, and we use it to construct an EL ratio CI in the simulation study in Section 5. To avoid estimating the scaling constant, we explore proper bootstrap procedures in Section 4.

4 Bootstrap EL intervals

We now propose two bootstrap procedures to construct EL CIs on the population mean of h(y). First, draw a bootstrap sample of size n using simple random sampling with replacement from the sample quadruples $\{(y_i, d_i, z_i, \delta_i): i \in s\}$, and denote the bootstrap sample as $\{(y_{b,i}, d_{b,i}, z_{b,i}, \delta_{b,i}): i = 1, ..., n\}$. Second, perform the imputation introduced in Section 2 on the bootstrap sample. That is, if $\delta_{b,i} = 0$ for

some i = 1, ..., n and $z_{b,i} = k \in \{1, ..., K\}$, select J values at random with replacement from the bootstrap donor set of class k, $\mathcal{R}_{b,k} = \{(y_{b,i}, d_{b,i}): \delta_{b,i} = 1, z_{b,i} = k, i = 1, ..., n\}$, and denote these values as $y_{b,ij}^*$, j = 1, ..., J. Then, similar to (2.1), define a bootstrap version of the imputed estimating function, denoted $\tilde{h}_{b,i}(\mu)$, for all i = 1, ..., n as

$$\tilde{h}_{b,i}(\mu) = \sum_{k=1}^{K} \mathbb{1} \Big(z_{b,i} = k \Big) \bigg[\delta_{b,i} d_{b,i} \Big\{ h \Big(y_{b,i} \Big) - \mu \Big\} + \Big(\mathbb{1} - \delta_{b,i} \Big) J^{-1} \sum_{j=1}^{J} d_{b,ij}^* \Big\{ h \Big(y_{b,ij}^* \Big) - \mu \Big\} \bigg].$$

Finally, obtain a bootstrap version of the profile log-EL, denoted $l_{b,n}(\mu)$, by replacing $\tilde{h}_i(\mu)$ in (3.1) with $\tilde{h}_{b,i}(\mu)$, and define the bootstrap MELE as $\hat{\mu}_b = \operatorname{argmin}_{\mu} l_{b,n}(\mu)$ and the bootstrap EL ratio as $R_b(\mu) = -2l_{b,n}(\mu)$.

To construct bootstrap CIs for μ , we seek suitable bootstrap analogues of $\sqrt{n} (\hat{\mu} - \mu_N)$ and $R(\mu_N)$. In particular, we propose asymptotically correct bootstrap quantities based on $\hat{\mu}_b$ and $R_b(\mu)$ that approximate the distributions of $\sqrt{n} (\hat{\mu} - \mu_N)$ and $R(\mu_N)$. We will further show that the usual bootstrap analogues $\sqrt{n} (\hat{\mu}_b - \hat{\mu})$ and $R_b(\hat{\mu})$, suggested by Shao and Sitter (1996), are asymptotically incorrect for approximating the distributions of $\sqrt{n} (\hat{\mu} - \mu_N)$ and $R(\mu_N)$ under fractional imputation with fixed J.

The proposed bootstrap analogues of $\sqrt{n}(\hat{\mu} - \mu_N)$ and $R(\mu_N)$ rely on a quantity which we call *complete-data MELE* as defined below. Let $n_k = \sum_{i \in s} \mathbb{1}(z_i = k)$ and $r_k = \sum_{i \in s} \delta_i \mathbb{1}(z_i = k)$ for k = 1, ..., K. For all $i \in s$, define

$$\breve{h}_{i}(\mu) = \sum_{k=1}^{K} \frac{n_{k}}{r_{k}} \delta_{i} \mathbf{1}(z_{i}=k) d_{i} \{h(y_{i}) - \mu\}.$$

Note that $\tilde{h}_i(\mu)$ does not involve imputation. Similar to (3.1), we define a profile log-EL based on $\tilde{h}_i(\mu)$,

$$\breve{l}_n(\mu) = \sup_{\{q_i\}} \left\{ \sum_{i \in s} \log q_i \colon q_i \ge 0, \sum_{i \in s} q_i = 1, \sum_{i \in s} q_i \breve{h}_i(\mu) = 0 \right\}.$$

Again, the design weights d_i are included in the definition of $\tilde{h}_i(\mu)$, although $\tilde{l}_n(\mu)$ does not explicitly depends on them. We then define the complete-data MELE as $\tilde{\mu} = \operatorname{argmin}_{\mu} \tilde{l}_n(\mu)$. As the profile log-EL defined in (3.1), the maximum of $\tilde{l}_n(\mu)$ is attained when $q_i = 1/n$, and, as a consequence, $\tilde{\mu}$ is the solution to the equation $n^{-1}\sum_{i \in s} \tilde{h}_i(\mu) = 0$, which is simply given by

$$\tilde{\mu} = \frac{\sum_{i \in S} \sum_{k=1}^{K} (n_k / r_k) \delta_i \mathbf{1}(z_i = k) d_i h(y_i)}{\sum_{i \in S} \sum_{k=1}^{K} (n_k / r_k) \delta_i \mathbf{1}(z_i = k) d_i}.$$
(4.1)

The complete-data MELE μ plays an important role in constructing asymptotically correct bootstrap quantities, as shown by Theorem 3.

Theorem 3. Let \mathcal{F}_n denote the sample data $\{(y_i, z_i, \delta_i): i \in s\}$. Under the conditions of Theorem 2,

$$\sup_{t} \left| P\left\{ \sqrt{n} \left(\hat{\mu}_{b} - \breve{\mu} \right) \le t \, \middle| \, \mathcal{F}_{n} \right\} - P\left\{ \sqrt{n} \left(\hat{\mu} - \mu_{N} \right) \le t \right\} \right| = o_{p} \left(1 \right)$$

$$(4.2)$$

and

$$\sup \left| P\left\{ R_{b}\left(\breve{\mu}\right) \leq t \,\middle|\, \mathcal{F}_{n}\right\} - P\left\{ R\left(\mu_{N}\right) \leq t\right\} \right| = o_{p}\left(1\right). \tag{4.3}$$

The proof of Theorem 3 is given in Appendix B.

Remark 1. The difference between the usual bootstrap quantity $\sqrt{n}(\hat{\mu}_b - \hat{\mu})$ (or $R_b(\hat{\mu})$) and the proposed bootstrap quantity $\sqrt{n}(\hat{\mu}_b - \bar{\mu})$ (or $R_b(\bar{\mu})$) can be shown to be $O_p(1)$ instead of $o_p(1)$ when J is a fixed constant. This, together with Theorem 3, shows that the usual bootstrap quantities do not have the same limiting distributions as those of $\sqrt{n}(\hat{\mu} - \mu_N)$ and $R(\mu_N)$, and will lead to asymptotically incorrect coverage of μ_N . If J is allowed to increase to ∞ as $n \to \infty$, then the differences between the usual bootstrap quantities and the proposed quantities becomes $o_p(1)$ and both are asymptotically correct.

Two bootstrap approaches to constructing a $(1 - \alpha)$, $\alpha \in (0, 1)$, level CI on μ are suggested by Theorem 3. Independently generate b = 1, ..., B bootstrap samples, and obtain $\hat{\mu}_b$ and $R_b(\bar{\mu})$ for all b. The first approach is based on the bootstrap distribution of $\sqrt{n} \{\hat{\mu}_b - \bar{\mu}\}$. Find the $(1 - \alpha/2)^{\text{th}}$ and $(\alpha/2)^{\text{th}}$ sample quantiles, $\hat{\mu}_{b,1-\alpha/2}$ and $\hat{\mu}_{b,\alpha/2}$, of $\{\hat{\mu}_b: b = 1, ..., B\}$. An approximate $(1 - \alpha)$ level CI for μ is given by

$$(\hat{\mu} - (\hat{\mu}_{b,1-\alpha/2} - \breve{\mu}), \hat{\mu} - (\hat{\mu}_{b,\alpha/2} - \breve{\mu})).$$

We call the above CI the bootstrap-EL percentile (BELP) interval.

The second approach relies on the bootstrap distribution of the bootstrap EL ratio $R_b(\breve{\mu})$. Find the $(1-\alpha)^{\text{th}}$ sample quantile, denoted $R_{b,1-\alpha}(\breve{\mu})$, of $\{R_b(\breve{\mu}): b = 1, ..., B\}$. Then an approximate $(1-\alpha)$ level CI for μ based on $R_b(\breve{\mu})$ is given by the interval defined by

$$\left\{\mu: R\left(\mu\right) \leq R_{b,1-\alpha}\left(\breve{\mu}\right)\right\}.$$

We call this CI the bootstrap-EL ratio (BELR) interval.

5 Simulation study

We carried out simulation studies to compare the performance of the proposed BELP and BELR intervals to that of the usual bootstrap intervals based on $\sqrt{n}(\hat{\mu}_b - \hat{\mu})$ and $R_b(\hat{\mu})$. We will refer to the proposed proper intervals as propBELP and propBELR, and refer to the usual naive ones as naiveBELP and naivBELR. We also report the results for EL ratio intervals (SELR) with estimated scaling constant c_N based on the limiting distribution of the EL ratio established in Theorem 2.

To generate population data, we followed the simulation settings of Wu and Rao (2006) and used the model

$$y_i = \beta_0 + \beta_1 x_i + \tau \epsilon_i \tag{5.1}$$

for i = 1, ..., N, where $\beta_0 = \beta_1 = 1$, x_i were generated from an exponential distribution with rate 1, and ϵ_i were generated from $\chi_1^2 - 1$ distribution to have a zero mean. The x_i were used as the size measure for selecting PPS samples. The value of τ was chosen so that the correlation $\rho := \rho(y, x)$ between the variable of interest y and the size measure x reaches a certain level. The finite populations so generated were held fixed under repeated independent simulation runs. In each particular case of the simulation study, the number of simulation runs was set to be 10,000, and the number of bootstrap replications in each simulation run was set to B = 3,000. In Sections 5.1 and 5.2, we focus on constructing 95% CIs for the population function of y evaluated at given values.

5.1 Case 1: Single imputation class

We first considered the simple case where there is only one imputation class for the entire population. We set the population size to be N = 5,000, and drew PPS samples with replacement from the population generated from model (5.1). The τ value in model (5.1) was chosen such that $\rho = 0.3$. The sample size was set to n = 80 and 250, and for each sample size, we examined two settings of response probability, P = 0.4 and 0.8. For each combination of n and P, we used two settings for the number of draws in imputation, J = 1 and 5. Note that, unlike the original setting used by Wu and Rao (2006), where a constant was added to all the size measures to avoid extremely small values, we intentionally avoided adding any constant to the size measures to test the case where inclusion probabilities contain very small values and differ largely in size.

The coverage probabilities and average lengths of the proposed and naive bootstrap-EL intervals for the population mean are shown in Table 5.1. When J = 1, it is clear that in all the cases the proposed propBELR intervals have the most accurate coverage probabilities and shorter average lengths than the naiveBELR intervals. The naivBELR intervals show 1%-2% of over-coverage relative to the 95% nominal coverage. Both propBELP and naivBELP intervals perform much worse than the BELR intervals in terms of coverage probabilities than the naivBELP intervals. The propBELP and propBELP intervals can be shown to have exactly the same lengths, so their lengths are shown in a single column titled "BELP" under "Average Length" in Table 5.1 and also in the other tables. Given that both the propBELP and naivBELP intervals. Moreover, when sample size increases, all the BELP intervals show improved coverage probabilities, while the coverages of the BELR intervals are stable in terms of change in the sample size. The SELR intervals also show significant under-coverage, although they have slightly better coverage than the propBELP intervals are stable in terms of change in the sample size. The SELR intervals also show significant under-coverage, although they have slightly better coverage than the propBELP intervals.

Ţ		<i>P</i> .		Cove	erage proba	bility			Averag	ge length	
J	п	I ₁	SELR	naivBELP	propBELP	naivBELR	propBELR	SELR	BELP	naivBELR	propBELR
1	80	0.4	0.880	0.785	0.832	0.965	0.951	3.094	3.081	4.416	4.199
		0.8	0.897	0.828	0.863	0.961	0.948	2.404	2.331	3.200	3.062
	250	0.4	0.900	0.827	0.886	0.966	0.951	2.161	2.069	2.832	2.646
		0.8	0.913	0.862	0.904	0.960	0.947	1.617	1.558	2.068	1.972
5	80	0.4	0.874	0.804	0.816	0.946	0.944	3.317	2.864	4.339	4.277
		0.8	0.890	0.850	0.858	0.947	0.944	2.339	2.222	3.074	3.038
	250	0.4	0.901	0.864	0.877	0.947	0.945	2.286	1.925	2.792	2.742
		0.8	0.908	0.889	0.902	0.948	0.946	1.567	1.463	1.939	1.914

Table 5.195% CIs for the population mean: single imputation class case

When J is increased to 5, the differences between the naivBELR and propBELR intervals, and those between the naivBELP and propBELP intervals, become nearly negligible. This observation agrees with our theoretical finding given in Remark 1, that is, as J increases, the differences between the proposed intervals and naive intervals will diminish as $n \rightarrow \infty$. The average lengths of the propBELR intervals remain slightly shorter than those of the naivBELR intervals. The coverage probabilities of both the propBELR intervals and the SELR intervals do not change substantially as J increases from 1 to 5.

A striking observation is that the BELP intervals perform much worse than the BELR intervals. Unreported simulation studies suggest that this is likely due to the use of unequal probability sampling. When simple random sampling is used, the performance of the propBELP interval is found to be close to that of the propBELR interval, which is also observed by Cai et al. (2019). In addition, if a constant is added to the size measures to avoid extremely small values, we observed that the performance of the proposed propBELP interval increases greatly while that of the naivBELP increases slightly. This is further illustrated in the simulation study presented in Section 5.2. A clear advantage of the proposed BELR interval over the proposed BELP interval is that the performance of the BELR interval is not significantly affected by the variation in inclusion probabilities.

The above simulation results show that the naivBELR intervals have similar performance to the propBELR intervals with a slight over-coverage. Does this imply that the naivBELR interval is also asymptotically correct? To answer this question, we conducted a large-sample simulation study. In this study, we set the population size to be N = 25,000, and considered sample sizes n = 500, 1,000, 1,500, 2,000 and 3,000. The population data were generated from model (5.1) and PPS samples were drawn with replacement from the population repeatedly and independently. The response probability P was fixed at 0.8 and the number of random draws J in the imputation was set to 1.

The simulation results based on large samples are reported in Table 5.2. In all the cases, the coverage probabilities of the propBELR intervals are precisely 95%. However, the naivBELR intervals always exihibt 1%-1.5% of over-coverage regardless of how large the sample size is. This shows that the naivBELR intervals are asymptotically biased. The coverage probabilities of the propBELP intervals improve as the sample size increases, and when n is beyond 2,000, they are satisfactorily close to the nominal coverage of 95%. The naivBELP intervals, however, have lower than 90% coverage probabilities in all the cases, implying that they are asymptotically incorrect. The SELR intervals also improve as the sample size n

increases. However, they improve slower than the propBELP intervals: when n = 250, the SELR intervals have slightly better coverage (Table 5.1), but when n increases to 3,000, the coverage probability of the propBELP intervals becomes 94.5% while that of the SELR intervals is only 93.0%.

		Cov	erage probal	bility		Average length			
п	SELR	naivBELP	propBELP	naivBELR	propBELR	SELR	BELP	naivBELR	propBELR
500	0.918	0.871	0.921	0.963	0.950	1.269	1.200	1.626	1.549
1,000	0.922	0.886	0.933	0.963	0.951	0.965	0.909	1.222	1.164
1,500	0.925	0.890	0.939	0.962	0.949	0.825	0.773	1.039	0.987
2,000	0.926	0.892	0.943	0.964	0.950	0.741	0.693	0.939	0.894
3,000	0.930	0.896	0.945	0.963	0.949	0.624	0.585	0.788	0.749

Large-sample behaviour of the 95% CIs for the population mean: single imputation class case

5.2 Case 2: Multiple imputation classes

We now turn to the case of multiple imputation classes, i.e., K > 1. We still focus on constructing 95% CIs for the population mean of y. We drew Rao-Sampford (Rao, 1965; Sampford, 1967) PPS samples without replacement from a finite population generated from model (5.1). In this study, we added the constant 1 to all the size measures generated from the standard exponential distribution to avoid extremely small values. We considered two settings of the sample size and population size combinations: (a) n = 150 and N = 5,000, corresponding to a sampling fraction of 3%, and (b) n = 500 and N = 50,000, corresponding to a sampling fraction of 1%. The reason that we reduced the sampling fraction in setting (b) is that for the large sample size n = 500, the rejective Rao-Sampford PPS samples are difficult to generate when the sampling fraction is greater than 1%. For each sample size setting, we considered two levels of correlation between y and the size measure, $\rho = 0.3$ and 0.8. Under each of the above sample size and correlation setting, we tested three cases for the number of random draws in the imputation: J = 1, 3 and 5.

We set the number of imputation classes to K = 3, and use the models considered by Fang, Hong and Shao (2009) to generate the class variable z and response probabilities for different imputation classes. The class variable z was generated with a proportional-odds model for all population units i = 1, ..., N,

$$\log \frac{P(z_i \le k \mid y_i)}{P(z_i > k \mid y_i)} = k + by_i \text{ for } k = 1, ..., K - 1$$

with b = -0.2. For each sampled unit $i \in s$, the response probability for y_i was generated according to the model

$$P(\delta_i | z_i = k) = \frac{\exp(-0.1 + \gamma k)}{1 + \exp(-0.1 + \gamma k)}$$

with $\gamma = 0.7$, where k = 1, ..., K. This model yields response probabilities $P_1 = 0.646$, $P_2 = 0.786$, and $P_3 = 0.881$.

The simulation results for the sample size n = 150 are reported in Table 5.3. In all the cases, the propBELR intervals have the most accurate coverage probabilities and shorter average lengths than the

Table 5.2

naivBELR intervals. The naivBELR intervals show over-coverage when J = 1, and their coverages improve as J increases. The coverage probabilities of the propBELP intervals are lower than the nominal level, but are much improved compared to the serious under-coverage that we have observed in the simulation study of Section 5.1 where no constant was added to the size measures to avoid extremely small values. The SELR intervals exhibit slight under-coverage when $\rho = 0.3$; they perform equally well as the propBELR intervals when $\rho = 0.8$. The naivBELP intervals perform the worst with significant undercoverage in all the cases.

Table 5.3
95% CIs for the population mean: multiple imputation classes case $n = 150$, $N = 5,000$

ρ	T		Cov	erage probal	oility		Average length				
μ	J	SELR	naivBELP	propBELP	naivBELR	propBELR	SELR	BELP	naivBELR	propBELR	
0.3	1	0.943	0.890	0.930	0.961	0.950	1.379	1.332	1.517	1.439	
	3	0.945	0.916	0.930	0.954	0.951	1.335	1.281	1.426	1.397	
	5	0.946	0.919	0.930	0.953	0.950	1.327	1.271	1.410	1.390	
0.8	1	0.953	0.899	0.946	0.967	0.952	0.471	0.465	0.502	0.474	
	3	0.951	0.928	0.945	0.956	0.950	0.455	0.444	0.464	0.453	
	5	0.951	0.935	0.945	0.956	0.952	0.447	0.440	0.455	0.449	

The simulation results for the larger sample size n = 500 are shown in Table 5.4. As in the case of the smaller sample size n = 150, the propBELR and propBELP intervals outperform their naive counterparts, and the propBELR intervals perform better than the propBELP intervals in terms of coverage probability. Under both sample sizes, 150 and 500, the coverage probabilities of the propBELR intervals are nearly identical to the nominal level, and those of the propBELP intervals improve as the sample size increases, suggesting that the proposed BELR and BELP intervals are asymptotically correct. However, the coverage probabilities of the naivBELR and naivBELP intervals do not improve as the sample size increases, indicating that they are asymptotically incorrect. The SELR intervals have approximately the same performance as the propBELR intervals in terms of both coverage probabilities and average lengths under the large sample size setting.

	Ţ		С	overage probal	bility		Average length				
ρ	J	SELR	naivBELP	propBELP	naivBELR	propBELR	SELR	BELP	naivBELR	propBELR	
0.3	1	0.949	0.895	0.939	0.960	0.948	0.747	0.734	0.795	0.754	
	3	0.948	0.926	0.941	0.954	0.949	0.720	0.704	0.742	0.727	
	5	0.949	0.933	0.942	0.952	0.950	0.719	0.698	0.731	0.721	
0.8	1	0.949	0.898	0.947	0.964	0.949	0.260	0.258	0.276	0.260	
	3	0.949	0.931	0.949	0.957	0.952	0.248	0.246	0.254	0.248	
	5	0.950	0.937	0.946	0.954	0.949	0.246	0.244	0.250	0.246	

Table 5.4 95% CIs for the population mean: multiple imputation classes case n = 500, N = 50,000

It is worth noting that the average lengths of all the intervals are shorter when the correlation between y and the size measure is higher. This agrees with the classical estimation theory of survey sampling that

using a size measure that is highly correlated with the variable of interest leads to small variance of the estimator.

5.3 Case 3: CIs for population distribution function

We now present the simulation results on 95% CIs of the finite-population distribution function of y at a given value t, $F_N(t) = N^{-1} \sum_{i \in U} \mathbf{1}(y_i \le t)$. As noted in the Introduction, $F_n(t)$ can be represented as the solution to the estimating equation $\sum_{i \in U} (h(y_i) - \mu) = 0$ in μ by taking $h(y) = 1 (y \le t)$. We took the same settings for data generation as used in simulation Case 2 in Section 5.2. For each ρ value, we considered three t values fixed at the 25^{th} , 50^{th} and 75^{th} percentiles of the data-generating distribution of y implied by model (5.1). For $\rho = 0.3$, these t values are 0.81, 1.95 and 3.99, and for $\rho = 0.8$, they are 2.09, 2.68 and 3.56. The sample size was set to n = 80.

The simulation results for the cases $\rho = 0.3$ and $\rho = 0.8$ are shown in Table 5.5 and 5.6, respectively. Consistent to what we have observed in simulation Case 2, the propBELR intervals still perform the best among the competitors. The SELR intervals show slight under-coverage compared to the propBELR intervals when $\rho = 0.3$ at the 75th percentile (t = 3.99); otherwise they perform similarly. The naivBELR intervals again exhibit approximately 1% of over-coverage when J = 1 and improve as J increases. The propBELP intervals perform better than the naivBELP intervals, but both show significant under-coverage.

4	T		Cov	erage probal	oility			Averag	e length	
t	J	SELR	naivBELP	propBELP	naivBELR	propBELR	SELR	BELP	naivBELR	propBELR
0.81	1	0.945	0.862	0.910	0.961	0.948	0.267	0.273	0.286	0.271
	3	0.948	0.898	0.912	0.953	0.947	0.257	0.261	0.263	0.257
	5	0.946	0.898	0.909	0.949	0.947	0.254	0.258	0.258	0.255
1.95	1	0.950	0.887	0.934	0.966	0.952	0.285	0.292	0.303	0.287
	3	0.951	0.916	0.932	0.958	0.951	0.273	0.279	0.280	0.274
	5	0.952	0.923	0.933	0.956	0.953	0.270	0.277	0.274	0.271
3.99	1	0.946	0.873	0.920	0.962	0.950	0.233	0.236	0.248	0.236
	3	0.947	0.907	0.922	0.955	0.950	0.225	0.227	0.231	0.227
	5	0.948	0.914	0.921	0.954	0.951	0.223	0.225	0.228	0.225

naivBELR

0.280

0.274

0.228

0.211

0.208

0.280

0.277

0.217

0.207

0.206

0.273

0.270

0.215

0.206

0.204

propBELR

0.279 0.266

0.263

0.288

0.274

0.271

0.217

0.207

0.205

Table 5.5 95% CIs for distribution function $F_N(t)$ when $\rho = 0.3$

Table 5.6 95% CIs for distribution function $F_N(t)$ when $\rho = 0.8$

0.913

0.924

0.878

0.906

0.912

					-					
4	t I Coverage probability							Average length		
ı	J	SELR	naivBELP	propBELP	naivBELR	propBELR	SELR	BELP	naivBEL	
2.09	1	0.942	0.860	0.905	0.958	0.944	0.277	0.283	0.295	
	3	0.946	0.891	0.910	0.948	0.945	0.265	0.271	0.272	
	5	0.947	0.900	0.910	0.947	0.946	0.263	0.269	0.267	
2.68	1	0.944	0.882	0.928	0.960	0.946	0.285	0.292	0.304	

0.951

0.951

0.957

0.949

0.947

0.945

0.948

0.945

0.946

0.946

0.931

0.934

0.919

0.920

0.922

0.946

0.949

0.943

0.944

0.944

3

5

1

3

5

3.56

6 Conclusion and future perspectives

A fractional imputation is proposed and an associated EL is developed for making inference on the population mean μ of a function of a variable of interest for survey data that are missing at random under PPS sampling with negligible sampling fractions or with replacement. Two bootstrap and EL based methods, BELP and BELR, are proposed for constructing CIs on μ , and are shown to be asymptotically correct. Simulation studies show that the proposed intervals perform better than their naive bootstrap counterparts under various sample size settings. In addition, the proposed BELR intervals are seen to have more accurate coverage probabilities than those of the proposed BELP intervals, particularly in two situations: (i) when the inclusion probabilities differ in size substantially, or (ii) when the sample size is not too large. Moreover, the proposed BELR intervals exhibit notably better coverage probabilities than the EL-ratio intervals with estimated scaling constant (SELR intervals) in the above case (i).

For parameters defined by smooth estimating equations under an IID setting, Tang and Qin (2012), using an inverse-probability-weighted fractional imputation, achieved two most desirable properties of EL inference with data missing at random: (1) the associated MELE attains the semi-parametric efficiency bound, and (2) the corresponding EL ratio follows a simple chi-square limiting distribution. However, their method requires both the observed and the missing data to be imputed, which is unlikely to be accepted in practice in survey studies. We are working on extending Tang and Qin (2012) to a survey setup while avoiding imputing observed data points.

Our future work also includes extending the current methods to tackling missing data from stratified sampling and multistage sampling, as well as to the case where the sampling fraction is non-negligible.

Appendix A

Define, for all k = 1, ..., K,

$$Q_{k} = \sum_{i \in U} \mathbf{1}(z_{i} = k) (nd_{i})^{-1},$$

$$\overline{H}_{k} = N^{-1} \sum_{i \in U} \mathbf{1}(z_{i} = k) \{h(y_{i}) - \mu_{N}\},$$

and

$$S_{H_{k}}^{2} = N^{-1} \sum_{i \in U} \mathbf{1}(z_{i} = k) \{h(y_{i}) - \mu_{N}\}^{2} (nd_{i}/N) - \overline{H}_{k}^{2}/Q_{k}.$$

Theorems 1-3 are established under the following regularity conditions.

- (R.1) Q_k , \overline{H}_k and $S^2_{H_k}$ converge to some constant limits as $N \to \infty$ for all k = 1, ..., K; $Q_k \neq 0$ for all k and $S^2_{H_k} \neq 0$ for at least one k.
- (R.2) There exists a constant $\epsilon > 0$ such that (a) $N^{-1} \sum_{i \in U} |h(y_i) \mu_N|^{2+\epsilon} = O(1)$, and (b) $N^{-(1+\epsilon)} \sum_{i \in U} d_i^{\epsilon} = o(1)$ as $N \to \infty$.

(R.3) $\max_{i \in s} |d_i \{h(y_i) - \mu_N\}| = o_p(n^{1/2}).$

Conditions (R.1) and (R.2) ensure that the population has regular behaviour when embedded in a asymptotic sequence. Condition (R.3) is a standard condition in EL inference as used by Chen and Sitter (1999).

The constant σ_N in Theorem 1 is given by the positive square root of

$$\sigma_N^2 = \sum_{k=1}^{K} \left[\left\{ P_k^{-1} + (1 - P_k) J^{-1} \right\} S_{H_k}^2 + \overline{H}_k^2 / Q_k \right].$$

The constant c_N in Theorem 2 is given by

$$c_{N} = \frac{\sigma_{N}^{2}}{\sum_{k=1}^{K} \left[\{P_{k} + (1 - P_{k}) J^{-1} \} S_{H_{k}}^{2} + \overline{H}_{k}^{2} / Q_{k} \right]}.$$

Design-consistent estimators of σ_N^2 and c_N can be obtained by plugging in the following design-consistent estimators of P_k , Q_k , \overline{H}_K and $S_{H_k}^2$ for k = 1, ..., K:

$$\hat{P}_{k} = \frac{\sum_{i \in s} \delta_{i} \mathbf{1}(z_{i} = k) d_{i}}{\sum_{i \in s} \mathbf{1}(z_{i} = k) d_{i}}, \ \hat{Q}_{k} = n^{-1} \sum_{i \in s} \mathbf{1}(z_{i} = k),$$

$$\hat{H}_{k} = N^{-1} P_{k}^{-1} \sum_{i \in s} \delta_{i} \mathbf{1}(z_{i} = k) d_{i} (h(y_{i}) - \hat{\mu}),$$

$$\hat{S}_{H_{k}}^{2} = N^{-1} P_{k}^{-1} \sum_{i \in s} \delta_{i} \mathbf{1}(z_{i} = k) (h(y_{i}) - \hat{\mu})^{2} (nd_{i}^{2} / N) - \hat{Q}_{k}^{-1} \hat{H}_{k}^{2}.$$

Appendix B

We now give proofs for Theorems 1-3. Let $E(\cdot)$ and $Var(\cdot)$ denote the expectation and variance operators with respect to the sampling design, respectively. We consider PPS sampling with replacement in the proofs. All results also apply to PPS sampling without replacement with negligible sampling fractions.

Lemma 1. Under condition (R.1), $n_k / r_k = P_k^{-1} = o_p(1)$.

Proof of Lemma 1. Recall that $n_k = \sum_{i \in S} 1(z_i = k)$, so we have $E(n_k/n) = \sum_{i \in U} 1(z_i = k)(nd_i)^{-1} = Q_k = O(1)$ by (R.1). Under PPS sampling with replacement, we have

$$\operatorname{Var}(n_k/n) = n^{-1}\left\{\sum_{i \in U} \mathbb{1}(z_i = k) (nd_i)^{-1} - \mathbb{E}^2(n_k/n)\right\} = n^{-1}\left\{Q_k - Q_k^2\right\} = o(1)$$

Therefore, by Markov's inequality, we easily obtain $n_k / n = E(n_k / n) + o_p(1) = Q_k + o_p(1)$.

Moreover, since $r_k = \sum_{i \in s} \delta_i \mathbf{1}(z_i = k)$, by the MAR assumption, we have $\mathbf{E}(r_k/n) = \mathbf{E}(\delta_i | z_i = k) \mathbf{E}(n_k/n) = P_k \mathbf{E}(n_k/n) = P_k Q_k$. Similar to the case of n_k/n , we can show that $\operatorname{Var}(r_k/n) = o(1)$, so we have $r_k/n = P_k Q_k + o_p(1)$.

We hence conclude that $n_k / r_k = (n_k / n) / (r_k / n) = P_k^{-1} + o_p (1)$.

Lemma 2. Under conditions (R.1) and (R.2),

$$\sigma_{N}^{-1}\sqrt{n}N^{-1}\sum_{i\in s}\tilde{h}_{i}(\mu_{N}) \stackrel{d}{\rightarrow} N(0,1).$$

Proof of Lemma 2. The following decomposition holds:

$$\sqrt{n}N^{-1}\sum_{i\in s}\tilde{h}_i(\mu_N) = U_n + V_n,$$

where $U_n = \sqrt{n}N^{-1}\sum_{i\in s} \left[\tilde{h}_i(\mu_N) - \mathbf{E}\left\{\tilde{h}_i(\mu_N) \mid \mathcal{F}_n\right\}\right]$ and $V_n = \sqrt{n}N^{-1}\sum_{i\in s} \mathbf{E}\left\{\tilde{h}_i(\mu_N) \mid \mathcal{F}_n\right\}$. Noting that $\mathbf{E}\left\{\tilde{h}_i(\mu_N) \mid \mathcal{F}_n\right\} = \sum_{k=1}^{K} \mathbf{1}(z_i = k) \left[\delta_i d_i \left\{h(y_i) - \mu_N\right\} + (1 - \delta_i) r_k^{-1} \hat{H}_{kr}\right]$, where $\hat{H}_{kr} = \sum_{i\in s} \mathbf{1}(z_i = k) \delta_i d_i \left\{h(y_i) - \mu_N\right\}$, we get

$$U_n = \frac{\sqrt{n}}{N} \sum_{i \in S} \eta_i$$
 and $V_n = \frac{\sqrt{n}}{N} \sum_{k=1}^K \frac{n_k}{r_k} \hat{H}_{kr}$ (B.1)

with

$$\eta_i = \sum_{k=1}^{K} (1 - \delta_i) \mathbf{1} (z_i = k) J^{-1} \sum_{j=1}^{J} \left[d_{ij}^* \left\{ h \left(y_{ij}^* \right) - \mu_N \right\} - r_k^{-1} \hat{H}_{kr} \right].$$

We now work out the conditional limiting distributions of U_n given the sample data \mathcal{F}_n . Since the imputation is carried out independently across $i \in s$, η_i are conditionally independent random variables given \mathcal{F}_n . Note that η_i have zero conditional mean, $E(\eta_i | \mathcal{F}_n) = 0$ and common conditional variance for all $i \in s$. Moreover, we have

$$\operatorname{Var}(U_{n} | \mathcal{F}_{n}) = \sum_{k=1}^{K} \left(\frac{n_{k}}{r_{k}} - 1\right) \frac{1}{J} \left[\frac{n}{N^{2}} \sum_{i \in s} 1(z_{i} = k) \,\delta_{i} d_{i}^{2} \left\{h(y_{i}) - \mu_{N}\right\}^{2} - \frac{n}{r_{k}} \left(\frac{\hat{H}_{kr}}{N}\right)^{2}\right]$$

By Lemma 1, $n_k / r_k = P_k^{-1} + o_p(1)$, and in the proof of Lemma 1, we have shown $n/r_k = P_k^{-1}Q_k^{-1} + o_p(1)$. Under condition (R.2)(a), we can show that

$$\frac{n}{N^2} \sum_{i \in s} \mathbf{1}(z_i = k) \, \delta_i d_i^2 \left\{ h(y_i) - \mu_N \right\}^2 = \frac{1}{N} P_k \sum_{i \in s} \mathbf{1}(z_i = k) \left\{ h(y_i) - \mu_N \right\}^2 (nd_i/N) + o_p(1)$$

and $\hat{H}_{kr}/N = P_k \overline{H}_k + o_p$ (1). Therefore,

$$s_{u}^{2} := \operatorname{Var}\left(U_{n} \mid \mathcal{F}_{n}\right) = \sum_{k=1}^{K} (1 - P_{k}) J^{-1} S_{H_{k}}^{2} + o_{p}(1) = O_{p}(1).$$
(B.2)

By the Berry-Essen Theorem (Chow and Teicher, 1997, Section 9.1), we have

$$\sup_{t} |P(U_{n} \leq ts_{u} | \mathcal{F}_{n}) - \Phi(t)| \leq c_{\epsilon} (t_{u} / s_{u})^{2+\epsilon},$$

where $\epsilon' \in (0, \epsilon)$ and $c_{\epsilon'}$ are some constants that do not rely on *n*, and

$$\iota_{u}^{2+\epsilon'} = \sum_{i\in s} \mathbb{E}\left\{\left(\sqrt{n}N^{-1}\eta_{i}\right)^{2+\epsilon'} \middle| \mathcal{F}_{n}\right\} = \frac{n^{2+\epsilon'}/2}{N^{2+\epsilon'}} \left\{\frac{1}{n}\sum_{i\in s} \mathbb{E}\left(\eta_{i}^{2+\epsilon'} \middle| \mathcal{F}_{n}\right)\right\}.$$

We can further show that, under condition (R.2)(a), $n^{-1}\sum_{i \in s} \mathbb{E}(\eta_i^{2+\epsilon_i} | \mathcal{F}_n) = O_p(1)$, which implies that $\iota_u^{2+\epsilon_i} = O_p(1)$. Therefore,

$$\sup_{t} \left| P\left(U_{n} \leq ts_{u} \middle| \mathcal{F}_{n} \right) - \Phi\left(t \right) \right| = o_{p}\left(1 \right).$$
(B.3)

We next find the limiting distribution of V_n . It can be shown that, for each k = 1, ..., K,

$$\frac{1}{N}\frac{n_k}{r_k}\hat{H}_{kr} = \frac{1}{n}\sum_{i\in s}\mathbf{1}(z_i=k)\left[\frac{\delta_i}{P_k}\left\{\frac{n}{N}d_i(h(y_i)-\mu_N)-\frac{\bar{H}_k}{Q_k}\right\}+\frac{\bar{H}_k}{Q_k}\right]+o_p(n^{-1/2}).$$

Hence, by (B.1), we have $V_n = \frac{\sqrt{n}}{n} \sum_{i \in S} \zeta_i + o_p(1)$, where

$$\zeta_i = \sum_{k=1}^{K} \mathbf{1}(z_i = k) \left[\frac{\delta_i}{P_k} \left\{ \frac{n}{N} d_i \left(h(y_i) - \mu_N \right) - \frac{\overline{H}_k}{Q_k} \right\} + \frac{\overline{H}_k}{Q_k} \right].$$

Under sampling with replacement, ζ_i , $i \in s$, are independent random variables. Moreover, we get $E\left(\sum_{i \in s} \zeta_i\right) = 0$ and

$$s_{v}^{2} := \operatorname{Var}(V_{n}) = \sum_{k=1}^{K} \{P_{k}^{-1}S_{H_{k}}^{2} + \overline{H}_{k}^{2}/Q_{k}\}.$$
 (B.4)

Under condition (R.2)(a), the conditions of the Lyanunov central limit theorem are satisfied, so we have

$$s_{\nu}^{-1}V_{n} \xrightarrow{d} N(0,1).$$
 (B.5)

By (B.3) and (B.5), and observing that s_v/s_u converges to a constant limit under condition (R.1), all conditions of Theorem 2 of Chen and Rao (2007) are verified. Accordingly, we have

$$\left(s_{u}^{2}+s_{v}^{2}\right)^{-1/2}\left(U_{n}+V_{n}\right) \xrightarrow{d} N\left(0,1\right)$$

Noticing that $s_u^2 + s_v^2 = \sigma_N^2 + o_p(1)$ and $U_n + V_n = \sqrt{n}N^{-1}\sum_{i \in S} \tilde{h}_i(\mu_N)$, the claimed result is proved.

Proof of Theorem 1. By (3.3) and (2.1), we have

$$\sqrt{n}\left(\hat{\mu}-\mu_{N}\right) = \sqrt{n}\tilde{d}^{-1}\sum_{i\in s}\tilde{h}_{i}\left(\mu_{N}\right), \tag{B.6}$$

where $\tilde{d} = \sum_{i \in s} \sum_{k=1}^{K} \mathbf{1}(z_i = k) \{ \delta_i d_i + (1 - \delta_i) J^{-1} \sum_{j=1}^{J} d_{ij}^* \}.$

Note that, given the sample data \mathcal{F}_n , $\mathbb{E}(N^{-1}\tilde{d} | \mathcal{F}_n) = N^{-1} \sum_{k=1}^{K} (n_k / r_k) \sum_{i \in s} \mathbb{1}(z_i = k) \delta_i d_i$. By Lemma 1, we have $n_k / r_k = P_k^{-1} + o_p$ (1). Moreover, we can show that, under condition (R.2)(b),

$$N^{-1}\sum_{i\in s} \mathbf{1}(z_{i} = k) \,\delta_{i}d_{i} = \mathbf{E}\left\{N^{-1}\sum_{i\in s} \mathbf{1}(z_{i} = k) \,\delta_{i}d_{i}\right\} + o_{p}(1)$$
$$= N^{-1}P_{k}\sum_{i\in U} \mathbf{1}(z_{i} = k) + o_{p}(1).$$

Therefore,

$$\mathbb{E}\left(N^{-1}\tilde{d} \mid \mathcal{F}_{n}\right) = \sum_{k=1}^{K} P_{k}^{-1} \left\{N^{-1}P_{k}\sum_{i \in U} \mathbf{1}(z_{i} = k)\right\} + o_{p}(1) = 1 + o_{p}(1).$$

In addition, it can be shown that given \mathcal{F}_n , $N^{-1}\tilde{d} = \mathbb{E}(N^{-1}\tilde{d} | \mathcal{F}_n) + o_p(1)$. The above results imply that

$$N^{-1}\tilde{d} = 1 + o_p(1). \tag{B.7}$$

Combining (B.6), (B.7) and Lemma 2, we obtain the desired result.

Proof of Theorem 2. By (3.2) and (3.4), we have

$$R(\mu_N) = -2l_n(\mu_N) = 2\sum_{i\in s} \log\{1 + \lambda_N \tilde{h}_i(\mu_N)\}.$$
(B.8)

where λ_N satisfies $n^{-1} \sum_{i \in s} \tilde{h}_i(\mu_N) / \{1 + \lambda_N \tilde{h}_i(\mu_N)\} = 0$. Using the same argument as used by Owen (2001, Section 11.2, Proof of Theorem 3.2), we can show that, under condition (R.3), $\lambda_N = O_p(n^{-1/2})$ and

$$\lambda_{N} = \frac{\sum_{i \in s} \tilde{h}_{i}(\mu_{N})}{\sum_{i \in s} \tilde{h}_{i}^{2}(\mu_{N})} + o_{p}(n^{-1/2})$$

By this expression of λ_N , (B.8) and a Taylor's expansion, we obtain

$$R(\mu_{N}) = 2\lambda_{N}\sum_{i\in s}\tilde{h}_{i}(\mu_{N}) - \lambda_{N}^{2}\sum_{i\in s}\tilde{h}_{i}^{2}(\mu_{N}) + o_{p}(1)$$
$$= \frac{\left\{\sqrt{n}N^{-1}\sigma_{N}^{-1}\sum_{i\in s}\tilde{h}_{i}(\mu_{N})\right\}^{2}}{nN^{-2}\sum_{i\in s}\tilde{h}_{i}^{2}(\mu_{N})/\sigma_{N}^{2}} + o_{p}(1).$$
(B.9)

Note that $nN^{-2}\sum_{i\in s}\tilde{h}_{i}^{2}(\mu_{N})=A_{1}+A_{2}$, where

$$A_{1} = n^{-1} \sum_{i \in S} \sum_{k=1}^{K} \delta_{i} \mathbf{1}(z_{i} = k) \{h(y_{i}) - \mu_{N}\}^{2} (nd_{i}/N)^{2},$$

and

$$A_{2} = n^{-1} \sum_{i \in s} \sum_{k=1}^{K} (1 - \delta_{i}) \mathbf{1}(z_{i} = k) J^{-2} \left[\sum_{j=1}^{J} \left\{ h(y_{ij}^{*}) - \mu_{N} \right\} (nd_{ij}^{*}/N) \right]^{2}.$$

Under condition (R.2)(a), we can show that $A_1 = \sum_{k=1}^{K} P_k \left(S_{H_k}^2 + \overline{H}_k^2 / Q_k \right) + o_p (1)$ and $A_2 = \sum_{k=1}^{K} (1 - P_k) \left(J^{-1} S_{H_k}^2 + \overline{H}_k^2 / Q_k \right) + o_p (1)$. Hence, $nN^{-2} \sum_{i \neq j} \tilde{h}_i^2 (\mu_N) = \sum_{k=1}^{K} \left[\{ P_k + (1 - P_k) J^{-1} \} S_{H_k}^2 + \overline{H}_k^2 / Q_k \right] + o_p (1).$ Theorem 2 is then proved by substituting the above expression into (B.9) and applying Theorem 1.

For the proof of Theorem 3, we introduce the following Lemma.

Lemma 3. Under conditions (R.1) and (R.2),

$$\sup_{t} \left| P\left(\sigma_{N}^{-1}\sqrt{n}N^{-1}\left\{\sum_{i=1}^{n}\tilde{h}_{b,i}\left(\mu_{N}\right)-\breve{h}_{+}\left(\mu_{N}\right)\right\} \leq t \left|\mathcal{F}_{n}\right]-\Phi\left(t\right) \right| = o_{p}\left(1\right)$$

where $\breve{h}_{+}(\mu_{N}) = \sum_{i \in s} \breve{h}_{i}(\mu_{N}).$

Proof of Lemma 3. Let $\mathcal{G}_{b,n}$ denote the bootstrap sample $\{(y_{b,i}, d_{b,i}, z_{b,i}, \delta_{b,i}): i = 1, ..., n\}$. We have the following decomposition:

$$\sqrt{n}N^{-1}\sum_{i=1}^{n}\tilde{h}_{b,i}(\mu_N) = \mathbb{U}_n + \mathbb{V}_n,$$

where $\mathbb{U}_{n} = \sqrt{n}N^{-1}\sum_{i=1}^{n} \left[\tilde{h}_{b,i}\left(\mu_{N}\right) - \mathrm{E}\left\{\tilde{h}_{b,i}\left(\mu_{N}\right) \middle| \mathcal{G}_{b,n}\right\}\right]$ and

$$\mathbb{V}_{n} = \sqrt{n} N^{-1} \left[\sum_{i=1}^{n} \mathrm{E} \left\{ \tilde{h}_{b,i} \left(\mu_{N} \right) \middle| \mathcal{G}_{b,n} \right\} - \breve{h}_{+} \left(\mu_{N} \right) \right].$$

Similar to the proof of Lemma 2, we can show that

$$\sup_{t} \left| P\left(\mathbb{U}_{n} \leq t s_{u} \left| \mathcal{G}_{b,n} \right) - \Phi\left(t\right) \right| = o_{p}\left(1\right), \tag{B.10}$$

where s_u is defined in (B.2).

We next give the conditional limiting distribution of \mathbb{V}_n given \mathcal{F}_n . It can be shown that $\mathbb{V}_n = \sqrt{n}N^{-1}\sum_{i=1}^n \xi_{b,i} + o_p(1)$, where

$$\xi_{b,i} = \sum_{k=1}^{K} \Big[P_k^{-1} \delta_{b,i} \mathbf{1} \big(z_{b,i} = k \big) \Big\{ d_{b,i} \big(h \big(y_{b,i} \big) - \mu_N \big) - r_k^{-1} \hat{H}_{kr} \Big\} + \mathbf{1} \big(z_{b,i} = k \big) r_k^{-1} \hat{H}_{kr} \Big] - n^{-1} \breve{h}_+ \big(\mu_N \big).$$

Conditioned on \mathcal{F}_n , $\xi_{b,i}$ are IID across i = 1, ..., n, and we can show that $E(\xi_{b,i} | \mathcal{F}_n) = 0$ and

$$\operatorname{Var}\left(\frac{\sqrt{n}}{N}\sum_{i=1}^{n}\xi_{b,i} \mid \mathcal{F}_{n}\right) = \sum_{k=1}^{K}P_{k}^{-2}\frac{n}{N^{2}}\sum_{i\in s}\delta_{i}\mathbf{1}(z_{i}=k)\left\{d_{i}\left(h\left(y_{i}\right)-\mu_{N}\right)-\frac{1}{r_{k}}\hat{H}_{kr}\right\}^{2} + \sum_{k=1}^{K}\frac{n}{N^{2}}n_{k}\left\{\frac{1}{r_{k}}\hat{H}_{kr}\right\}^{2}.$$

We can show that the first term on the right hand side (RHS) equals $\sum_{k=1}^{K} P_k^{-1} S_{H_k}^2 + o_p(1)$ and the second term on the RHS equals $\sum_{k=1}^{K} \overline{H}_k^2 / Q_k + o_p(1)$. Therefore

$$\operatorname{Var}\left(\sqrt{n}N^{-1}\sum_{i=1}^{n}\xi_{b,i} \mid \mathcal{F}_{n}\right) = \sum_{k=1}^{K} \left\{ P_{k}^{-1}S_{H_{k}}^{2} + \overline{H}_{k}^{2} / Q_{k} \right\} + o_{p}\left(1\right) = s_{v}^{2} + o_{p}\left(1\right),$$

where s_{ν}^2 is defined in (B.4). By verifying the conditions of the Berry-Essen Theorem as in the proof of Lemma 2, we get

$$\sup_{t} |P(\mathbb{V}_{n} \leq ts_{v} | \mathcal{F}_{n}) - \Phi(t)| = o_{p}(1).$$
(B.11)

By (B.10) and (B.11), and applying Theorem 2 of Chen and Rao (2007), we obtain the claimed results.

Proof of Theorem 3. We first prove (4.2). By (3.3) and (2.1), we have

$$\sqrt{n} \left(\hat{\mu}_b - \mu_N \right) = \sqrt{n} \tilde{d}_b^{-1} \sum_{i=1}^n \tilde{h}_{b,i} \left(\mu_N \right),$$

where $\tilde{d}_{b} = \sum_{i=1}^{n} \sum_{k=1}^{K} \mathbf{1} (z_{b,i} = k) \{ \delta_{b,i} d_{b,i} + (1 - \delta_{b,i}) J^{-1} \sum_{j=1}^{J} d_{b,ij}^{*} \}$. By (4.1), we have

$$\sqrt{n}\left(\breve{\mu}-\mu_{N}\right)=\sqrt{n}\breve{d}^{-1}\sum_{i\in s}\breve{h}_{i}\left(\mu_{N}\right)=\sqrt{n}\breve{d}^{-1}\breve{h}_{+}\left(\mu_{N}\right),$$

where $\vec{d} = \sum_{i \in s} \sum_{k=1}^{K} (n_k / r_k) \mathbf{1}(z_i = k) \delta_i d_i$.

It is straightforward to show that $\tilde{d}_b/N = \tilde{d}/N + o_p(1)$ and $\tilde{d}/N = 1 + o_p(1)$ under condition (R.2)(b). Therefore,

$$\sqrt{n}N^{-1}\left(\hat{\mu}_{b}-\breve{\mu}\right)=\sqrt{n}N^{-1}\left\{\sum_{i=1}^{n}\tilde{h}_{b,i}\left(\mu_{N}\right)-\breve{h}_{+}\left(\mu_{N}\right)\right\}+o_{p}\left(1\right).$$

Then, by Lemma 3, we have

$$\sup_{t} \left| P\left(\sqrt{n} N^{-1} \left(\hat{\mu}_{b} - \breve{\mu} \right) \leq t \sigma_{N} \left| \mathcal{F}_{n} \right) - \Phi(t) \right| = o_{p}(1).$$

This, combined with Theorem 1 and Polya's Theorem, completes the proof of (4.2).

The result (4.3) can be proved based on (4.2) and by following the same arguments as used in the proof of Theorem 2.

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Improved Horvitz-Thompson estimator in survey sampling

Xianpeng Zong, Rong Zhu and Guohua Zou¹

Abstract

The Horvitz-Thompson (HT) estimator is widely used in survey sampling. However, the variance of the HT estimator becomes large when the inclusion probabilities are highly heterogeneous. To overcome this shortcoming, in this paper we propose a hard-threshold method for the first-order inclusion probabilities. Specifically, we carefully choose a threshold value, then replace the inclusion probabilities smaller than the threshold by the threshold. Through this shrinkage strategy, we construct a new estimator called the improved Horvitz-Thompson (IHT) estimator to estimate the population total. The IHT estimator increases the estimation accuracy much although it brings a bias which is relatively small. We derive the IHT estimator's mean squared error and its unbiased estimator, and theoretically compare the IHT estimator with the HT estimator. We also apply our idea to construct an improved ratio estimator. We numerically analyze simulated and real data sets to illustrate that the proposed estimators are more efficient and robust than the classical estimators.

Key Words: Horvitz-Thompson estimator; Inverse probability weighting; Hard-threshold; Robustness; Unequal probability sampling; Sampling without/with replacement; Ratio estimator.

1 Introduction

The Horvitz-Thompson (HT) estimator proposed by Horvitz and Thompson (1952) is widely used in survey sampling. It has also been applied to other fields such as functional data analysis (Cardot and Josserand, 2011) and the treatment effect (Rosenbaum, 2002). The HT estimator is an unbiased estimator constructed via inverse probability weighting. However, when the inclusion probabilities are highly heterogeneous, i.e., inclusion probabilities of some units are relatively tiny, the variance of the HT estimator becomes large due to inverse probability weighting. In this paper, we propose an improved Horvitz-Thompson (IHT) estimator to address this problem.

Our approach is to use a hard-threshold for the first-order inclusion probabilities. Specifically, we carefully choose an inclusion probability as the threshold. The inclusion probabilities that are smaller than the threshold are replaced by the threshold, while the others remain unchanged. In this way, we obtain the modified inclusion probabilities, and construct an estimator based on the modified inclusion probabilities through inverse probability weighting. We call this estimator the IHT estimator. This method looks very easy but is more efficient than the HT estimator. This hard-threshold approach can be explained as a shrinkage method. Shrinkage is very commonly used in statistics, such as ridge regression (Hoerl and Kennard, 1970) and high-dimensional statistics (Tibshirani, 1996). In this paper, we use it to reduce the negative effect of highly heterogeneous inclusion probabilities. Similar to other shrinkage methods, our approach introduces a bias, which is proved to be very small, but reduces the variance to a larger extent, so it improves the estimation efficiency. We will theoretically and numerically show the improvement from using the modified inclusion probabilities. In addition to the population total estimator, we also extend this strategy to the ratio estimator, and accordingly, an improved ratio estimator is obtained.

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The remainder of the paper is organized as follows. Section 2 introduces the HT estimator and shows its drawback. Section 3 proposes our modified inclusion probabilities and the resultant IHT estimator. We also provide the IHT estimator's properties, and theoretically compare it with the HT estimator in this section. Section 4 extends our idea to obtain an improved ratio estimator and shows that this modification is efficient. Section 5 presents numerical evidence from simulations and a real data analysis. Section 6 concludes. Proofs of theoretical results are given in the Appendix.

2 HT estimator and its drawback

Consider a finite population $U = \{U_1, ..., U_N\}$ of size N, where U_k denotes the k^{th} unit. For simplicity, we write $U = \{1, ..., k, ..., N\}$. For each unit k, suppose that the value y_k of the target characteristic Y is measured. Our aim is to estimate the total, $t_y = \sum_U y_k$, using a sample s of size n which is randomly drawn from the population U. We implement unequal probability sampling without replacement. Denote $\{\pi_k\}_{k=1}^N$ as the first-order inclusion probabilities and $\{\pi_{kl}\}_{k\neq l}$ as the second-order inclusion probabilities.

Horvitz and Thompson (1952) proposed the HT estimator as follows

$$\hat{t}_{\rm HT} = \sum_{k \in s} \frac{y_k}{\pi_k}.$$

The HT estimator \hat{t}_{HT} is an unbiased estimator of t_y and its variance is

$$V(\hat{t}_{\rm HT}) = \sum_{U} \frac{\Delta_{kk}}{\pi_k^2} y_k^2 + \sum_{k \neq l} \frac{\Delta_{kl}}{\pi_k \pi_l} y_l y_k, \qquad (2.1)$$

where $\Delta_{kk} = \pi_k - \pi_k^2$ for all k and $\Delta_{kl} = \pi_{kl} - \pi_k \pi_l$ for all $k \neq l$. When the inclusion probabilities are highly imbalanced, i.e., some π_k 's are very small, the variance of the HT estimator may be very large.

3 Improved HT estimator

In this section, we improve the HT estimator in the sense of reducing its mean squared error (MSE). The resultant estimator is referenced as the IHT estimator. For doing this, we first propose the modified first-order inclusion probabilities, where the hard-threshold method is used to reduce the effect of those inclusion probabilities with relatively tiny values.

Definition 1. Let $\pi_{(1)} \leq \pi_{(2)} \leq ... \leq \pi_{(N)}$ be the ordered values of the first-oder inclusion probabilities $\{\pi_1, \pi_2, ..., \pi_N\}$. Assume that there exists an integer $K \geq 2$ such that $\pi_{(K)} \leq (K+1)^{-1}$. We define the modified first-order inclusion probabilities as follows

$$\pi_{k}^{*} = \begin{cases} \pi_{k} & \pi_{k} > \pi_{(K)}, \\ \pi_{(K)} & \pi_{k} \le \pi_{(K)}, \end{cases} \quad 1 \le k \le N.$$

From the definition, we partition the finite population into two parts: $U_1 = \{k : \pi_k > \pi_{(K)}\}$ with size N - K, and $U_2 = \{k : \pi_k \le \pi_{(K)}\}$ with size K. For U_1 , the first-order inclusion probabilities remain unchanged, while all of first-order inclusion probabilities for U_2 are replaced by $\pi_{(K)}$. From this hard-threshold, we get our modified first-order inclusion probabilities $\{\pi_k^*\}_{k=1}^N$. Obviously, the choice of K is very important. In Section 3.2, we shall provide a simple way to choose K.

Remark on existence of K. The assumption in Definition 1 is quite weak. If $\pi_{(2)} > 1/(2+1)$, then the sampling fraction $f > \frac{1}{3} - \frac{1}{3N}$. However that situation that $f > \frac{1}{3}$ rarely happens in practical surveys. Thus, the inequality that $\pi_{(2)} \le 1/(2+1)$ generally holds.

Instead of the original first-order inclusion probabilities $\{\pi_k\}_{k=1}^N$, we use our defined modified firstorder inclusion probabilities $\{\pi_k^*\}_{k=1}^N$ to construct an improved Horvitz-Thompson (IHT) estimator by inverse probability weighting.

Definition 2. The IHT estimator is defined as

$$\hat{t}_{IHT} = \sum_{k \in s} \frac{y_k}{\pi_k^*}.$$

Unlike the unbiased HT estimator, the IHT estimator is biased. However, this modification leads to much smaller MSE due to reducing the variance. It is worth pointing out that, although we focus on sampling without replacement in this paper, our modification idea is equally applicable to the Hansen-Hurwitz estimator (Hansen and Hurwitz, 1943) for sampling with replacement.

3.1 Properties of the IHT estimator

In this section, we derive the properties of the IHT estimator. We first provide the expressions of its bias, variance, MSE and an unbiased estimator of MSE in Theorem 1. Then we compare the IHT estimator with the HT estimator in Theorems 2 and 3.

Theorem 1. The bias and variance of the IHT estimator \hat{t}_{IHT} are expressed as

$$Bias(\hat{t}_{IHT}) = \sum_{U_2} \left(\frac{\pi_k}{\pi_{(K)}} - 1 \right) y_k,$$

and

$$Var(\hat{t}_{IHT}) = \sum_{U} \frac{\Delta_{kk}}{\pi_k^{*2}} y_k^2 + \sum_{k \neq l} \frac{\Delta_{kl}}{\pi_k^* \pi_l^*} y_k y_l,$$

respectively, where $\Delta_{kk} = \pi_k (1 - \pi_k)$, $\Delta_{kl} = \pi_{kl} - \pi_k \pi_l (k \neq l)$ as defined before. Therefore, its MSE is given by

$$MSE(\hat{t}_{IHT}) = \left[\sum_{U_2} \left(\frac{\pi_k}{\pi_{(K)}} - 1\right) y_k\right]^2 + \sum_{U} \frac{\Delta_{kk}}{\pi_k^{*2}} y_k^2 + \sum_{k \neq l} \frac{\Delta_{kl}}{\pi_k^* \pi_l^*} y_k y_l.$$
(3.1)

An unbiased estimator of the MSE is

$$\widehat{MSE}(\hat{t}_{IHT}) = \sum_{s_2} \frac{(\pi_k - \pi_{(K)})^2}{\pi_{(K)}^2 \pi_k} y_k^2 + \sum_{k \neq l} \sum_{s_2} \frac{(\pi_k - \pi_{(K)})(\pi_l - \pi_{(K)})}{\pi_{(K)}^2 \pi_{kl}} y_k y_l + \sum_s \frac{\breve{\Delta}_{kk}}{\pi_k^{*2}} y_k^2 + \sum_{k \neq l} \sum_s \frac{\breve{\Delta}_{kl}}{\pi_k^* \pi_l^*} y_k y_l,$$

where $\breve{\Delta}_{kk} = \frac{\Delta_{kk}}{\pi_k}$, $\breve{\Delta}_{kl} = \frac{\Delta_{kl}}{\pi_{kl}}$, *s* is the sample set, and $s_2 = s \cap U_2$.

Proof. See Appendix A.1.

To derive the properties of the IHT estimator, we need the following regularity conditions: **Condition C.1.** $\min_{i \in U} \pi_i \ge \lambda > 0$, $\min_{i,j \in U} \pi_{ij} \ge \lambda^* > 0$, and

$$\limsup_{Narrow\infty} n \max_{i\neq j\in U} \left| \pi_{ij} - \pi_i \pi_j \right| < \infty.$$

Condition C.2. $\max_{i \in U} |y_i| \le C$ with *C* a positive constant not depending on *N*.

Condition C.1 is a common condition imposed on the first-order and second-order inclusion probabilities. The same conditions are used in Breidt and Opsomer (2000), where further comments on C.1 are provided. Condition C.2 is also a common condition.

Theorem 2. For the HT estimator \hat{t}_{HT} and the IHT estimator \hat{t}_{IHT} , under the Conditions C.1-C.2, we have

$$Bias(N^{-1}\hat{t}_{HT}) = 0, \qquad Bias(N^{-1}\hat{t}_{HT}) = O(n^{-1});$$

and

$$MSE(N^{-1}\hat{t}_{HT}) = O(n^{-1}), \qquad MSE(N^{-1}\hat{t}_{HT}) = O(n^{-1}).$$

Proof. See Appendix A.2.

From Theorem 2, the squared-bias of our IHT estimator is very small compared to its MSE. Although our IHT estimator brings a bias to reduce the variance, the price for this is relatively small. The following theorem theoretically compares the efficiency of the two estimators.

Theorem 3. Under the Conditions C.1-C.2, we have

$$MSE\left(N^{-1}\hat{t}_{IHT}\right) \le MSE\left(N^{-1}\hat{t}_{HT}\right) + o\left(n^{-1}\right).$$
(3.2)

Especially, for Poisson sampling, we obtain

$$MSE\left(N^{-1}\hat{t}_{IHT}\right) \leq MSE\left(N^{-1}\hat{t}_{HT}\right),$$

where the strict inequality is true if there exist $k \neq l \in U_2$ such that $(\pi_k - \pi_{(K)}) y_k \neq (\pi_l - \pi_{(K)}) y_l$.

Proof. See Appendix A.3.

Theorem 3 shows that, under some mild conditions, the proposed IHT estimator is asymptotically more efficient than the HT estimator. From the proof in Appendix A.3, the term $o(n^{-1})$ in equation (3.2) is due to the interaction term from the second-order inclusion probabilities. We theoretically bound the term as $o(n^{-1})$. For Poisson sampling, the term does not exist, so the MSE of the IHT estimator is uniformly not larger than that of the HT estimator. Empirically, we compare the IHT estimator with the HT estimator in Section 5.

3.2 The choice of K

The efficiency of the IHT estimator relies on the choice of K, which provides a control of the varianceand-bias tradeoff. The choice of K needs to satisfy the condition that $\pi_{(K)} < 1/(K+1)$ of Definition 1, since the modified inclusion probabilities would cause large bias when K becomes large. On the other hand, the improvement of the IHT estimator would not be significant if K is small. In the proofs of Theorem 3, equation (A.5) provides a lower bound of the main term of $MSE(N^{-1}\hat{t}_{HT}) - MSE(N^{-1}\hat{t}_{IHT})$. The lower bound increases as $\pi_{(K)}$ increases. Therefore, denoting the maximum value $K^* = \max\{i: \pi_{(i)} \le 1/(i+1)\}$, we choose K^* as the threshold. In practice, we propose the following algorithm to find the maximum value K^* .

Algorithm 1	The choice of K
Step (i)	Obtain the ordered inclusion probabilities $\{\pi_{(1)}, \pi_{(2)},, \pi_{(N)}\}$ by sorting
	$\{\pi_k\}_{k=1}^N$ from small to large.
Step (ii)	Test and modify.
	If j satisfies $\pi_{(j)} \leq \frac{1}{j+1}$ and $\pi_{(j+1)} > \frac{1}{j+2}$, the modified first-order inclusion
	probabilities are defined as
	$oldsymbol{\pi}^* = igg\{\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!\!$
	and $K = j$.

Note that the choice of K^* based on Algorithm 1 is not optimal in terms of MSE. However, we simulate an example in Section 5 where the performance of Algorithm 1 is very close to that of the theoretically ideal choice.

4 Extension to the ratio estimator

When an auxiliary variable is available, the ratio estimator is usually used to estimate the population total. In this section, we extend the IHT estimator to the case of ratio estimation.

4.1 Improved ratio estimator

Denote the ratio between the population totals of Y and Z as

$$R = t_v / t_z$$
,

where t_y and t_z are the totals of the finite populations Y and Z, respectively. Let $\hat{t}_{y\pi} = \sum_{s} \frac{y_k}{\pi_k}$, $\hat{t}_{z\pi} = \sum_{s} \frac{z_k}{\pi_k}$, $\hat{t}_{y\pi}^* = \sum_{s} \frac{y_k}{\pi_k^*}$, and $\hat{t}_{z\pi}^* = \sum_{s} \frac{z_k}{\pi_k^*}$. The classical estimator and our modified estimator of R are given by

$$\hat{R} = \hat{t}_{y\pi} / \hat{t}_{z\pi}$$
, and $\hat{R}^* = \hat{t}_{y\pi}^* / \hat{t}_{z\pi}^*$.

We assume that the population total t_z is known. To estimate the population total t_y of Y, the classical ratio estimator is given by

$$\hat{Y}_{R} = t_{z} \cdot \hat{t}_{y\pi} / \hat{t}_{z\pi}$$

Alternatively, our improved ratio estimator of t_y based on the modified inclusion probabilities is expressed as

$$\hat{Y}_{R}^{*} = t_{z} \cdot \hat{t}_{y\pi}^{*} / \hat{t}_{z\pi}^{*}$$

4.2 Properties of the improved ratio estimator

To show theoretically that the improved ratio estimator \hat{Y}_{R}^{*} is more efficient than the classical ratio estimator \hat{Y}_{R} , we need the following regularity conditions:

Condition C.3. $\lim_{N\to\infty} \frac{n}{N} = c$, where $c \in (0, 1)$ is a constant.

Condition C.4. $\max_{i \neq j \neq k \in U} \left(\pi_{ijk} - \pi_{ij} \pi_k \right) = O(n^{-1}), and$

$$\max_{i\neq j\neq k\neq l\in U} \left(\pi_{ijkl} - 4\pi_{ijk}\pi_{l} + 6\pi_{ij}\pi_{k}\pi_{l} - 3\pi_{i}\pi_{j}\pi_{k}\pi_{l} \right) = O(n^{-2}).$$

Condition C.3 is a common condition. The same condition is used in Breidt and Opsomer (2000). Condition C.4 is a mild assumption on the third-order and fourth-order inclusion probabilities. In Appendix A.5, we present some frequent examples which satisfy Condition C.4.

Comparing our improved estimators with the classical estimators, we have the following result.

Theorem 4. If Conditions C.1-C.4 are satisfied, and $c_1 \le z_k \le c_2$ for all $k \in U$ with c_1 and c_2 some positive constants, then

$$MSE\left(\hat{R}^{*}\right) \leq MSE\left(\hat{R}\right) + o\left(n^{-1}\right).$$

Furthermore,

$$MSE\left(N^{-1}\hat{Y}_{R}^{*}\right) \leq MSE\left(N^{-1}\hat{Y}_{R}\right) + o\left(n^{-1}\right).$$

Proof. See Appendix A.4.

Like Theorem 3, Theorem 4 shows that the proposed method improves the classical ratio estimators with a tolerance of order $o(n^{-1})$.

5 Numerical studies

In this section, we assess the empirical performance of our IHT estimator using three synthetic examples and one real example. We consider the following two cases: the estimation of a population total and the estimation of a population ratio, where our IHT estimators are compared with the HT estimator. We measure the efficiency improvement in terms of $Re = \frac{|MSE^{HT} - MSE^{HT}|}{MSE^{HT}} \times 100\%$, where MSE^{HT} and MSE^{IHT} denote the MSE of the HT estimators and IHT estimators, respectively. We additionally compare the IHT estimator with the HT estimator in the sense of inference performance in the real example.

5.1 Simulations

Example 1: An illustrative example

We generate a finite population Y of size N = 3,000, where the k^{th} unit value $y_k = |y_{0k}|$ and $y_{0k} \sim N(0, 1)$. Our aim is to estimate the population mean $\overline{Y} = \frac{1}{N} \sum_U y_k$. We perform Poisson sampling according to the inclusion probabilities set as follows

$$\pi_1 = \ldots = \pi_{1,000} = 0.2, \ \pi_{1,001} = \ldots = \pi_{2,000} = 0.001, \ \text{and} \ \pi_{2,001} = \ldots = \pi_{3,000} = 0.08.$$

In this example, the HT estimator is not efficient since one third of the inclusion probabilities are 0.001, tiny relative to 0.08 or 0.2. From our hard-threshold strategy, we replace these tiny probabilities with 0.08, so the modified inclusion probabilities are given by

$$\pi_1^* = \ldots = \pi_{1,000}^* = 0.2, \ \pi_{1,001}^* = \ldots = \pi_{2,000}^* = 0.08, \ \text{and} \ \pi_{2,001}^* = \ldots = \pi_{3,000}^* = 0.08.$$

Note that the modified probabilities are not obtained according to Algorithm 1. It is an illustrative example to show that our hard-threshold can bring efficiency improvement. By setting the iteration time M = 2,000, we get the simulated biases, variances and MSEs of our IHT estimator and the HT estimator. The results are shown in Table 5.1.

Table 5.1Performance of Example 1

MSE ^{ht}	MSE ^{IHT}	Bias ^{HT}	Bias ^{IHT}	Var ^{HT}	Var ^{IHT}	Re
0.1187	0.0751	5.374×10-6	0.0723	0.1187	0.0029	36.71%

From the table, the variance of the HT estimator is much larger than that of the IHT estimator, so it loses its efficiency in terms of MSE compared to the IHT estimator although the HT estimator is unbiased. Furthermore, in order to show the variations of both estimators, we plot their values among 2,000 iterations

in Figure 5.1. It clearly displays that, although there is small bias for the IHT estimator, its variation is much less than that of the HT estimator. These observations empirically verify our theoretical results.

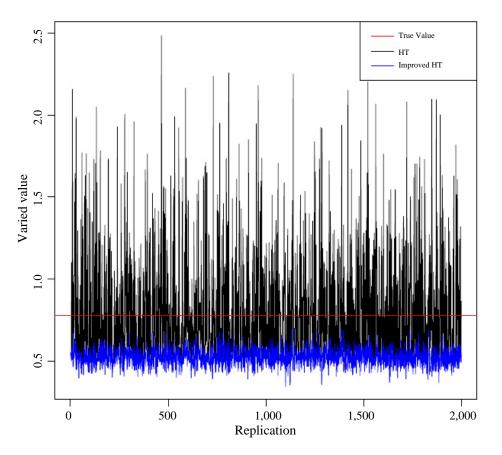


Figure 5.1 The plots of both estimators in Example 1.

Example 2: π_i 's depend on an auxiliary variable

We generate the finite population Y of size N = 3,000 as follows: $y_k = \sqrt{3} \cdot \rho \cdot x_k + \sqrt{3 - 3\rho^2} \cdot |e_k|$, where x_k and e_k are independently generated from U(0, 2) and N(0, 1) respectively, and $0 \le \rho \le 1$ controling the correlation of Y and X. We consider three sampling methods: Poisson sampling, PPS sampling and π PS sampling. The sampling fraction $f = \frac{n}{N} = 0.02, 0.04, 0.06, 0.08, 0.10, 0.15, 0.20, 0.30$. We report the results in Figure 5.2, where $\rho = 0.8$, and list the specific Re values of Figure 5.2 in Table 5.4.

From these results, we get the same observations as Example 1. It indicates that our IHT estimator outperforms the HT estimator in terms of MSE and that the improvement is generally substantial. Comparing with Figures 5.2(a), 5.2(b), and 5.2(c), π PS sampling obtains the biggest advantage of the IHT estimator over the HT estimator, followed by PPS sampling and Poisson sampling. We also show the results for different ρ values under π PS sampling in Table 5.2, where the case of f = 0.08 is reported and other cases are ignored because of the similarity. It is observed from the table that, no matter what value ρ takes, the IHT estimator has uniformly much less MSE than the HT estimator.

Table 5.2

The performance of Example 2 for different ρ values, where $f = 0.08$										
ρ	MSE ^{ht}	MSE ^{IHT}	Bias ^{HT}	Bias ^{IHT}	Var ^{HT}	Var ^{IHT}	Re			
0	3.45×10-2	1.36×10-2	3.43×10-5	5.82×10^{-4}	3.45×10^{-2}	1.30×10^{-2}	60.70%			
0.1	2.51×10-2	1.38×10^{-2}	1.16×10-5	8.25×10^{-4}	2.51×10^{-2}	1.30×10^{-2}	44.91%			
0.3	2.43×10-2	1.24×10^{-2}	4.65×10-6	8.86×10^{-4}	2.43×10^{-2}	1.15×10^{-2}	48.97%			
0.5	2.38×10-2	1.07×10^{-2}	9.83×10-6	8.44×10^{-4}	2.38×10^{-2}	9.88×10-3	54.92%			
0.8	9.38×10-3	5.22×10-3	3.04×10 ⁻⁷	3.16×10^{-4}	9.38×10-3	4.91×10-3	44.33%			
0.9	4.75×10-3	2.65×10-3	7.98×10-6	2.64×10^{-4}	4.74×10^{-3}	2.38×10^{-3}	44.27%			

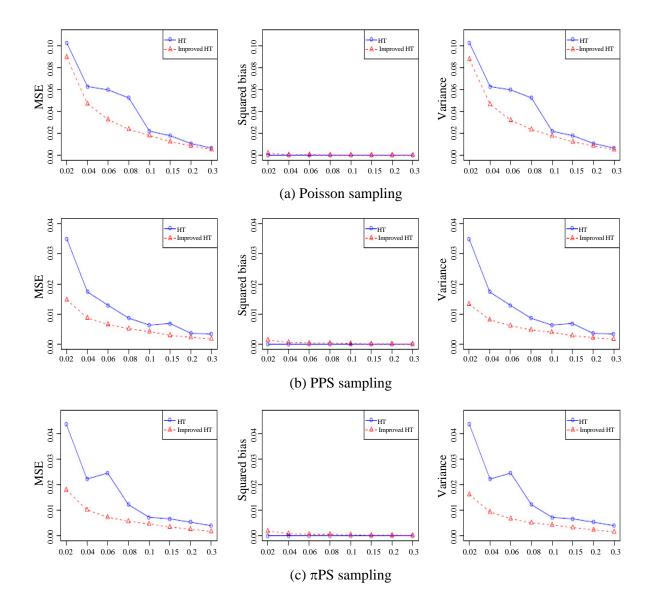


Figure 5.2 The performance of our IHT estimator and the HT estimator in Example 2, where $\rho = 0.8$. From left to right: the MSE performance, the squared-bias performance, and the variance performance.

Example 2 (continued): The performance of Algorithm 1 and the effect of the outcome's coefficient of variation

Here we empirically investigate the performance of Algorithm 1 and the effect of the outcome's coefficient of variation on our IHT estimator. We generate a finite population through a linear model with an intercept: $y_k = \alpha + \sqrt{3} \cdot \rho \cdot x_k + \sqrt{3-3\rho^2} \cdot |e_k|$, where x_k and e_k are the same as Example 2. We set N = 1,000, and control the coefficient of variation of the outcome by varying the intercept term $\alpha = \{-10, 5, 0, 5, 10\}$. Firstly, we study the performance of Algorithm 1. Note that the optimal choice K_{opt} can be derived via minimizing equation (3.1). We compare the MSE values based on K_{opt} and K^* from Algorithm 1, and report the results of f = 0.03 in Table 5.3 and ignore other cases because of the similarity. From the table, the MSE values based on K^* are very close to those based on K_{opt} . It indicates that Algorithm 1 provides an efficient choice of K. Secondly, we investigate the effect of the outcome's coefficient of variation. From the table, the IHT estimator always performs much better than the HT estimator when α takes different values. It indicates that our IHT is robust to the outcome's coefficient of variation.

Table 5.3Performance of Algorithm 1, where f = 0.03

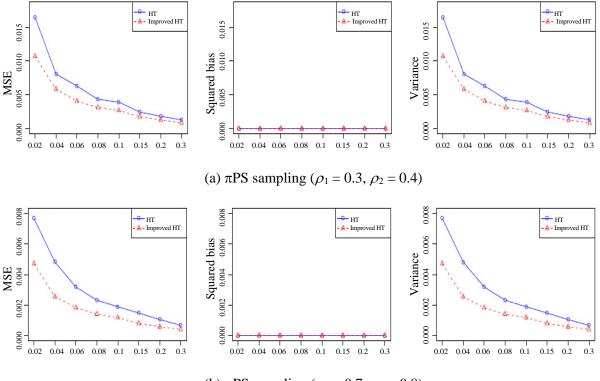
α	\overline{Y}	K *	K _{opt}	MSE _{ht}	MSE _{k*}	MSE opt	Re
-10	-7.80	125	166	3.3928	1.4130	1.3448	58.35%
-5	-2.81	125	174	0.7097	0.3073	0.2907	56.70%
0	2.19	125	164	0.0623	0.0245	0.0237	60.67%
5	7.20	125	160	1.4056	0.5884	0.5647	58.14%
10	12.24	125	159	4.7510	1.9916	1.9121	58.08%

Example 3: The estimation of population ratio

We generate two populations Y and Z of size N = 3,000: $y_k = \sqrt{12} \cdot \rho_1 \cdot x_k + \sqrt{3 - 3\rho_1^2} \cdot |e_1|$, and $z_k = \sqrt{12} \cdot \rho_2 \cdot x_k + \sqrt{3 - 3\rho_2^2} \cdot |e_2|$, where $x_k \sim U(0, 1)$, $e_1 \sim N(0, 1)$ and $e_2 \sim N(0, 1)$. Our aim is to estimate the ratio $R = t_y/t_z$, where $t_y = \sum_{k=1}^N y_k$ and $t_z = \sum_{k=1}^N z_k$. We set (ρ_1, ρ_2) as (0.3, 0.4) or (0.7, 0.8), and report the results of two cases in Figures 5.3(a) and 5.3(b), respectively. Similar to the estimation of the population total in examples given above, Figure 5.3 shows that our improved estimator outperforms the classical estimator. We also list the specific Re values of Figure 5.3 in Table 5.4, where the MSEs decrease by 27% to 47%.

Table 5.4Some specific Re values of Figures 5.2 and 5.3

f	0.02	0.04	0.06	0.08	0.10	0.15	0.20	0.30
Figure 5.2(a)	12.73%	25.33%	45.52%	54.71%	18.15%	30.94%	18.96%	21.99%
Figure 5.2(b)	57.92%	49.78%	49.48%	40.52%	33.81%	57.44%	36.45%	48.70%
Figure 5.2(c)	58.98%	54.41%	70.42%	53.75%	36.05%	48.72%	52.05%	57.65%
Figure 5.3(a)	35.09%	27.92%	35.16%	28.09%	31.50%	28.00%	29.07%	36.31%
Figure 5.3(b)	38.57%	47.18%	42.76%	39.27%	37.49%	46.20%	44.14%	39.55%



(b) π PS sampling ($\rho_1 = 0.7, \rho_2 = 0.8$)

Figure 5.3 Performance of Example 3. From left to right: the MSE performance, the squared-bias performance, and the variance performance.

5.2 Real example

We investigate the data set "Lucy" in the R package "TeachingSampling" (Gutierrez, 2009). This data set includes the variables of 2,396 firms: *ID*, *Level*, *Income*, *Employees*, and *Taxes*. Our aim is to estimate the *Employees* mean \overline{Y} of the 2,300 small or mid-sized firms ($\overline{Y} = 60.59$). We set the *Income* as the size of the firm, and perform π PS sampling. The sample size *n* is set among {46, 92, 138, 184, 230, 345, 460, 690}. We list the results in Table 5.5, where the bias, variance, MSE and Re values are reported. We also present the number K^* chosen by Algorithm 1. From Table 5.5, our IHT estimator has much better performance than the HT estimator in terms of MSE. As the sampling fraction *f* increases, the value of K^* decreases. It means that the number of the modified inclusion probabilities becomes weak when the sample size increases.

In this real example, we additionally compare the IHT estimator with the HT estimator in the sense of inference performance. Since the squared bias of the IHT estimator is negligible as shown in Theorem 2, the confidence region with 95% coverage is constructed as follow:

$$\left(\hat{t} - 1.96\sqrt{\widehat{\text{MSE}}}, \,\hat{t} + 1.96\sqrt{\widehat{\text{MSE}}}\right),$$
(5.1)

where \hat{t} is the IHT estimator, and \widehat{MSE} is its MSE estimator.

n	46	92	138	184	230	345	460	690
$MSE^{\rm HT}$	42.60	20.80	26.87	9.30	6.97	8.01	6.40	2.99
MSE ^{IHT}	28.27	14.05	10.18	7.75	5.70	3.77	2.85	1.76
Bias ^{HT}	0.0092	0.0002	0.0004	0.0020	0.0041	0.0001	0.0005	0.0112
Bias ^{IHT}	0.7520	0.3375	0.2562	0.1093	0.1253	0.0831	0.0539	0.0626
Var ^{HT}	42.59	20.80	26.87	9.30	6.97	8.01	6.40	2.97
Var ^{IHT}	27.52	13.71	9.92	7.64	5.57	3.68	2.79	1.70
Re ↑	33.64%	32.46%	62.13%	16.75%	18.31%	53.01%	55.49%	41.09%
K^*	166	100	72	59	49	36	29	21

Table 5.5The performance of estimation for the real data set "Lucy"

We iteratively simulate M = 5,000 times and calculate the mean and variance of MSE estimator, and the 95% coverage probabilities. The coverage probabilities (CP) are calculated as CP = $\frac{1}{M} \sum_{m=1}^{M} I(\overline{t} \in A_{(m)})$ where \overline{t} is the finite population mean and $A_{(m)}$ is the constructed 95% confidence region of the m^{th} iteration using equation (5.1). The inference performance is reported in Table 5.6. From the table, we have two observations. Firstly, our IHT estimator has smaller MSE than the HT estimator, but it attains almost the same coverage as the HT estimator. Thus, much narrower confidence intervals of the IHT estimator are constructed than those of the HT estimator. Secondly, for the HT estimator, the MSE estimator is much unstable due to the high heterogeneousness of the inclusion probabilities, while our IHT can efficiently overcome this problem. As a summary, our IHT estimator not only increases the estimator than the HT estimator.

 Table 5.6

 The inference performance of "Lucy" data set

f		ł	łT		IHT			
	MSE	$E(\widehat{MSE})$	Var (MSE)	СР	MSE	$E(\widehat{MSE})$	Var (MSE)	СР
0.02	219	76.1	8.28×10^{4}	91%	48.9	48.4	1.37×103	90%
0.04	109	173	2.90×107	92%	26.9	26.9	196	92%
0.06	72.7	118	9.11×106	91%	18.4	18.2	117	91%
0.08	54.3	67.5	1.94×106	93%	14.2	14.1	37.9	92%
0.10	43.2	59.5	1.46×106	93%	11.4	11.2	22.8	93%
0.15	28.5	27.2	2.40×105	93%	7.47	7.40	17.1	93%

6 Concluding remarks

In this paper, we have proposed a novel and simple method to improve the Horvitz-Thompson estimator in survey sampling. Compared with the HT estimator, the proposed IHT estimator improves the estimation accuracy at the expense of introducing a small bias. Empirical studies show that the improvement is substantial. This new idea has also been used to construct an improved ratio estimator. Naturally, applying it to other estimators, such as the regression estimator and the treatment effect estimator, is of interest as well, and this warrants further study.

The choice of the threshold K is important in our method. Although we have suggested an easy algorithm for the choice and have numerically showed that our choice is very close to the optimal one in terms of MSE, it may not be optimal in terms of MSE. How to choose an optimal threshold is a meaningful topic for future research.

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Appendix

A.1 Proof of Theorem 1

To obtain the MSE of the IHT estimator, we first define $I_k = 1$ or 0, k = 1, ..., N, if the k^{th} unit is drawn or not, then

$$E(I_k) = \pi_k$$
, $Var(I_k) = \Delta_{kk}$, $Cov(I_k, I_l) = \Delta_{kl}$ for $k \neq l$,

where $\Delta_{kk} = \pi_k (1 - \pi_k)$, $\Delta_{kl} = \pi_{kl} - \pi_k \pi_l$. So the bias of the IHT estimator is

$$\operatorname{Bias}\left(\hat{t}_{\operatorname{IHT}}\right) = E\left(\sum_{U} \frac{y_{k}}{\pi_{k}^{*}} I_{k}\right) - \sum_{U} y_{k} = \sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1\right) y_{k}.$$
(A.1)

The variance of the IHT estimator is given by

$$\operatorname{Var}\left(\hat{t}_{\mathrm{IHT}}\right) = \operatorname{Var}\left(\sum_{s} \frac{y_{k}}{\pi_{k}^{*}}\right) = \operatorname{Var}\left(\sum_{U} \frac{y_{k}}{\pi_{k}^{*}}I_{k}\right)$$
$$= \sum_{U} \left[\left(\frac{y_{k}}{\pi_{k}^{*}}\right)^{2} \operatorname{Var}\left(I_{k}\right)\right] + \sum_{k \neq l} \sum_{U} \left(\frac{y_{k}}{\pi_{k}^{*}} \frac{y_{l}}{\pi_{l}^{*}} \operatorname{Cov}\left(I_{k}, I_{l}\right)\right)$$
$$= \sum_{U_{1}} \frac{\Delta_{kk}}{\pi_{k}^{2}} y_{k}^{2} + \sum_{U_{2}} \frac{\Delta_{kk}}{\pi_{(K)}^{2}} y_{k}^{2} + \sum_{k \neq l} \sum_{U} \frac{\Delta_{kl}}{\pi_{k}^{*} \pi_{l}^{*}} y_{k} y_{l}.$$
(A.2)

Combining (A.1) and (A.2), we obtain

$$MSE(\hat{t}_{IHT}) = Bias^{2}(\hat{t}_{IHT}) + Var(\hat{t}_{IHT})$$

$$= \left[\sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1 \right) y_{k} \right]^{2} + \sum_{U} \frac{\Delta_{kk}}{\pi_{k}^{*2}} y_{k}^{2} + \sum_{k \neq l} \sum_{U} \frac{\Delta_{kl}}{\pi_{k}^{*} \pi_{l}^{*}} y_{k} y_{l}$$

$$= \left\{ \sum_{U} \frac{\Delta_{kk}}{\pi_{k}^{*2}} y_{k}^{2} + \left[\sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1 \right) y_{k} \right]^{2} \right\} + \sum_{k \neq l} \sum_{U} \frac{\Delta_{kl}}{\pi_{k}^{*} \pi_{l}^{*}} y_{k} y_{l}$$

$$\triangleq F_{1} + F_{2}. \qquad (A.3)$$

It is directly verified that $E\left(\widehat{\text{MSE}}\left(\hat{t}_{\text{IHT}}\right)\right) = \text{MSE}\left(\hat{t}_{\text{IHT}}\right)$. Therefore, Theorem 1 is proved.

A.2 Proof of Theorem 2

Using Conditions C.1 and C.2, we see that $\lambda \leq \pi_k \leq \pi_{(K)} \leq 1$ for each $k \in U_2$, and $\max_{k \neq l \in U_2} |\pi_{kl} - \pi_k \pi_l| = O(n^{-1})$. Then, from equation (2.1), we have

$$\begin{split} \left| E\left(\hat{\overline{t}}_{\mathrm{HT}} - \overline{t}\right)^{2} \right| &= \left| \frac{1}{N^{2}} \sum_{U} \frac{\Delta_{kk}}{\pi_{k}^{2}} y_{k}^{2} + \frac{1}{N^{2}} \sum_{k \neq l} \sum_{U} \frac{\Delta_{kl}}{\pi_{k} \pi_{l}} y_{l} y_{k} \right| \\ &\leq \frac{1}{N^{2}} \sum_{U} \frac{1 - \pi_{k}}{\pi_{k}} y_{k}^{2} + \frac{1}{N^{2}} \sum_{k \neq l} \sum_{U} \left| \frac{\pi_{kl} - \pi_{k} \pi_{l}}{\pi_{k} \pi_{l}} \right| |y_{l} y_{k}| \\ &= O\left(n^{-1}\right). \end{split}$$

Similarly, by the MSE of the IHT estimator given in (3.1), we observe

$$\begin{split} \left| E\left(\hat{t}_{\text{IHT}} - \overline{t}\right)^{2} \right| &= \left| \left[\frac{1}{N} \sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1 \right) y_{k} \right]^{2} + \frac{1}{N^{2}} \sum_{U} \frac{\Delta_{kk}}{\pi_{k}^{*2}} y_{k}^{2} + \frac{1}{N^{2}} \sum_{k \neq l} \sum_{U} \frac{\Delta_{kl}}{\pi_{k}^{*} \pi_{l}^{*}} y_{k} y_{l} \right| \\ &\leq \left[\frac{K}{N} \frac{1}{K} \sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1 \right) y_{k} \right]^{2} + \frac{1}{N^{2}} \sum_{U} \left| \frac{\pi_{k} \left(1 - \pi_{k} \right)}{\pi_{k}^{*2}} \right| y_{k}^{2} \\ &+ \frac{1}{N^{2}} \sum_{k \neq l} \left| \frac{\pi_{kl} - \pi_{k} \pi_{l}}{\pi_{k}^{*} \pi_{l}^{*}} \right| |y_{k} y_{l}| \\ &= O\left(n^{-1}\right). \end{split}$$

From Conditions C.1 and C.2, it is readily seen that

$$\operatorname{Bias}\left(\widehat{\overline{t}}_{\operatorname{IHT}}\right) = \left|\frac{1}{N}\sum_{U_{2}}\left(\frac{\pi_{k}}{\pi_{(K)}} - 1\right)y_{k}\right| \leq \frac{K}{N}\frac{1}{K}\sum_{U_{2}}\left|\frac{\pi_{k}}{\pi_{(K)}} - 1\right| |y_{k}| \leq \frac{1}{N}\sum_{U_{2}}|y_{k}| = O(n^{-1}),$$

where the third and fourth steps are valid due to $\lambda \leq \pi_k \leq \pi_{(K)} \leq 1$ for each $k \in U_2$ and $K/N = O(n^{-1})$, respectively.

A.3 Proof of Theorem 3

From equation (2.1), since the HT estimator is unbiased, we have

$$MSE(\hat{Y}_{HT}) = \left\{ \sum_{U_1} \frac{\Delta_{kk}}{\pi_k^2} y_k^2 + \sum_{U_2} \frac{\Delta_{kk}}{\pi_k^2} y_k^2 \right\} + \sum_{k \neq l} \sum_{U_1} \frac{\Delta_{kl}}{\pi_k \pi_l} y_k y_l \triangleq F_3 + F_4.$$
(A.4)

To illustrate the effectiveness of the new estimator, we compare equation (A.3) and equation (A.4). We prove $F_3 \ge F_1$ at first. It is clear that

$$F_{3} - F_{1} = \sum_{U} \frac{\Delta_{kk}}{\pi_{k}^{2}} y_{k}^{2} - \left\{ \sum_{U_{1}} \frac{\Delta_{kk}}{\pi_{k}^{2}} y_{k}^{2} + \sum_{U_{2}} \frac{\Delta_{kk}}{\pi_{(K)}^{2}} y_{k}^{2} + \left[\sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1 \right) y_{k} \right]^{2} \right\}$$
$$= \sum_{U_{2}} \frac{\Delta_{kk}}{\pi_{k}^{2}} y_{k}^{2} - \sum_{U_{2}} \frac{\Delta_{kk}}{\pi_{(K)}^{2}} y_{k}^{2} - \left[\sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1 \right) y_{k} \right]^{2}$$
$$= \sum_{U_{2}} \frac{\left(\pi_{(K)}^{2} - \pi_{k}^{2} \right) (1 - \pi_{k})}{\pi_{(K)}^{2} \pi_{k}} y_{k}^{2} - \left[\sum_{U_{2}} \left(\frac{\pi_{k}}{\pi_{(K)}} - 1 \right) y_{k} \right]^{2}$$
$$\triangleq D - C.$$

Using the Cauchy-Schwarz inequality, we have

$$C = \left(\sum_{U_2} \frac{\pi_k - \pi_{(K)}}{\pi_{(K)}} y_k\right)^2 \le K \sum_{U_2} \frac{\left(\pi_k - \pi_{(K)}\right)^2}{\pi_{(K)}^2} y_k^2 \triangleq E,$$

where the strict inequality holds if there exist $k \neq l \in U_2$ such that $(\pi_k - \pi_{(K)}) y_k \neq (\pi_l - \pi_{(K)}) y_l$. Further,

$$F_{3} - F_{1} \ge D - E = \sum_{U_{2}} \frac{\left(\pi_{(K)}^{2} - \pi_{k}^{2}\right)(1 - \pi_{k})}{\pi_{(K)}^{2}\pi_{k}} y_{k}^{2} - K \sum_{U_{2}} \frac{\left(\pi_{k} - \pi_{(K)}\right)^{2}}{\pi_{(K)}^{2}} y_{k}^{2}$$
$$= \sum_{U_{2}} \frac{\left(\pi_{(K)} - \pi_{k}\right)\left[\left(1 - \pi_{k} - K\pi_{k}\right)\pi_{(K)} + \left(\pi_{k} - \pi_{k}^{2} + K\pi_{k}^{2}\right)\right]}{\pi_{(K)}^{2}\pi_{k}} y_{k}^{2}.$$
(A.5)

From Definition 1, we have $\pi_k \le \pi_{(K)} \le (K+1)^{-1}$ for each $k \in U_2$, thus $D - E \ge 0$. So $F_3 - F_1 = D - C \ge D - E \ge 0$ holds.

For the terms F_2 and F_4 , we note that

$$\begin{split} F_2 - F_4 &= \sum_{k \neq l} \sum_{U} \frac{\Delta_{kl}}{\pi_k^* \pi_l^*} y_k y_l - \sum_{k \neq l} \sum_{U} \frac{\Delta_{kl}}{\pi_k \pi_l} y_k y_l \\ &= \sum_{k \neq l} \sum_{U_2} \left(\frac{\Delta_{kl}}{\pi_{(K)}^2} - \frac{\Delta_{kl}}{\pi_k \pi_l} \right) y_k y_l + \sum_{k \in U_1} \sum_{l \in U_2} \left(\frac{\Delta_{kl}}{\pi_{(K)} \pi_k} - \frac{\Delta_{kl}}{\pi_k \pi_l} \right) y_k y_l \\ &+ \sum_{k \in U_2} \sum_{l \in U_1} \left(\frac{\Delta_{kl}}{\pi_{(K)} \pi_l} - \frac{\Delta_{kl}}{\pi_k \pi_l} \right) y_k y_l \\ &\triangleq \Delta_1 + \Delta_2 + \Delta_3. \end{split}$$

Using Conditions C.1 and C.2, it is seen that

$$\begin{aligned} \frac{|\Delta_{1}|}{N^{2}} &= \frac{1}{N^{2}} \left| \sum_{k \neq l} \sum_{U_{2}} \frac{\pi_{kl} - \pi_{k} \pi_{l}}{\pi_{k} \pi_{l}} \left(\frac{\pi_{k} \pi_{l}}{\pi_{(K)}^{2}} - 1 \right) y_{k} y_{l} \right| \\ &\leq \frac{1}{N^{2}} \sum_{k \neq l} \sum_{U_{2}} \left| \frac{\pi_{kl} - \pi_{k} \pi_{l}}{\pi_{k} \pi_{l}} \right| \left| \frac{\pi_{k} \pi_{l}}{\pi_{(K)}^{2}} - 1 \right| |y_{k} y_{l}| \\ &\leq \frac{K^{2}}{N^{2}} \frac{1}{K^{2}} \sum_{k \neq l} \sum_{U_{2}} \left| \frac{\pi_{kl} - \pi_{k} \pi_{l}}{\pi_{k} \pi_{l}} \right| \left| |y_{k} y_{l}| = O(n^{-3}), \end{aligned}$$

where the third and fourth steps are valid due to $\lambda \leq \pi_k \leq \pi_{(K)} \leq 1$ for each $k \in U_2$, $K/N = O(n^{-1})$, and $\max_{k \neq l \in U_2} |\pi_{kl} - \pi_k \pi_l| = O(n^{-1})$. Similarly, we obtain

$$\begin{aligned} \frac{|\Delta_2|}{N^2} &= \frac{1}{N^2} \left| \sum_{k \in U_1 l \in U_2} \frac{\pi_{kl} - \pi_k \pi_l}{\pi_k \pi_l} \left(\frac{\pi_l}{\pi_{(K)}} - 1 \right) y_k y_l \right| \\ &\leq \frac{1}{N^2} \sum_{k \in U_1 l \in U_2} \left| \frac{\pi_{kl} - \pi_k \pi_l}{\pi_k \pi_l} \right| \left| \frac{\pi_l}{\pi_{(K)}} - 1 \right| |y_k y_l| \\ &\leq \frac{1}{N^2} \sum_{k \in U_1 l \in U_2} \left| \frac{\pi_{kl} - \pi_k \pi_l}{\pi_k \pi_l} \right| |y_k y_l| = O(n^{-2}), \end{aligned}$$

and $\frac{|\Delta_3|}{N^2} = O(n^{-2}).$

Thus, together with $F_3 \ge F_1$, we have

$$\mathrm{MSE}\left(N^{-1}\hat{t}_{\mathrm{IHT}}\right) \leq \mathrm{MSE}\left(N^{-1}\hat{t}_{\mathrm{HT}}\right) + o\left(n^{-1}\right)$$

For the Poisson sampling case, we have $F_4 = F_2 = 0$. Hence, for Poisson sampling, we obtain

$$\mathrm{MSE}\left(N^{-1}\hat{t}_{\mathrm{IHT}}\right) \leq \mathrm{MSE}\left(N^{-1}\hat{t}_{\mathrm{HT}}\right).$$

A.4 Proof of Theorem 4

First note that

$$\left(\hat{R}-R\right)^{2} = \left(\frac{\hat{t}_{y\pi}-R\hat{t}_{z\pi}}{\hat{t}_{z\pi}}\right)^{2} = \frac{\left(\hat{t}_{y\pi}-R\hat{t}_{z\pi}\right)^{2}}{t_{z}^{2}} - \frac{\left(\hat{t}_{z\pi}^{2}-t_{z}^{2}\right)\left(\hat{t}_{y\pi}-R\hat{t}_{z\pi}\right)^{2}}{t_{z}^{2}\hat{t}_{z\pi}^{2}} \triangleq \mathrm{I} + \mathrm{III},$$

and

$$\left(\hat{R}^* - R\right)^2 = \left(\frac{\hat{t}_{y\pi}^* - R\hat{t}_{z\pi}}{\hat{t}_{z\pi}^*}\right)^2 = \frac{\left(\hat{t}_{y\pi}^* - R\hat{t}_{z\pi}^*\right)^2}{t_z^2} - \frac{\left(\hat{t}_{z\pi}^{*2} - t_z^2\right)\left(\hat{t}_{y\pi}^* - R\hat{t}_{z\pi}^*\right)^2}{t_z^2 \hat{t}_{z\pi}^{*2}} \triangleq \text{II} + \text{IV}.$$

Let $u_k = y_k - Rz_k$. By Theorem 3, we have

$$N^{-2}E(\hat{t}_{u}^{*}-t_{u})^{2} \leq N^{-2}E(\hat{t}_{u}-t_{u})^{2}+o(n^{-1}).$$

Thus, for the terms I and II, we get

$$E(\mathbf{I}) \le E(\mathbf{II}) + o(n^{-1}). \tag{A.6}$$

Now, we need to prove that the expectations of III and IV are negligible. Observe that,

$$\begin{split} \left| E(\text{III}) \right| &= \left| E \frac{\left(\hat{t}_{z\pi} + t_z \right) \left(\hat{t}_{z\pi} - t_z \right) \left(\hat{t}_{y\pi} - R \hat{t}_{z\pi} \right)^2}{t_z^2 \, \hat{t}_{z\pi}^2} \right| \\ &\leq E \frac{\left| \hat{t}_{z\pi} + t_z \right| \left| \hat{t}_{z\pi} - t_z \right| \left(\hat{t}_{y\pi} - R \hat{t}_{z\pi} \right)^2}{t_z^2 \, \hat{t}_{z\pi}^2} \\ &\leq \frac{Z^* + \left| t_z \right|}{t_z^2 \, Z_*^2} E\left(\left| \hat{t}_{z\pi} - t_z \right| \left(\hat{t}_{y\pi} - R \hat{t}_{z\pi} \right)^2 \right) \\ &\leq \frac{Z^* + \left| t_z \right|}{t_z^2 \, Z_*^2} \sqrt{E\left(\hat{t}_{z\pi} - t_z \right)^2 E\left(\hat{t}_{y\pi} - R \hat{t}_{z\pi} \right)^4}, \end{split}$$

where $Z^* = \frac{n}{N} \max_{k \in U} \left(\frac{z_k}{\pi_k} \right)$, $Z_* = \frac{n}{N} \min_{k \in U} \left(\frac{z_k}{\pi_k} \right)$. Similarly,

$$|E(IV)| \leq \frac{\tilde{Z}^{*} + |t_{z}|}{t_{z}^{2}\tilde{Z}_{*}^{2}} \sqrt{E(\hat{t}_{z\pi}^{*} - t_{z})^{2}E(\hat{t}_{y\pi}^{*} - R\hat{t}_{z\pi}^{*})^{4}},$$

where $\tilde{Z}^* = \frac{n}{N} \max_{k \in U} \left(\frac{z_k}{\pi_k^*} \right), \ \tilde{Z}_* = \frac{n}{N} \min_{k \in U} \left(\frac{z_k}{\pi_k^*} \right).$

Using Theorem 2 and Lemma 1, we see that $|E(III)| = O(n^{-3/2})$ and $|E(IV)| = O(n^{-3/2})$. Combining these and equation (A.6), we get

$$\operatorname{MSE}\left(\hat{R}^{*}\right) \leq \operatorname{MSE}\left(\hat{R}\right) + o\left(n^{-1}\right).$$

It implies that $\operatorname{MSE}\left(N^{-1}\hat{Y}_{R}^{*}\right) \leq \operatorname{MSE}\left(N^{-1}\hat{Y}_{R}\right) + o\left(n^{-1}\right)$.

A.5 Discussion on Condition C.4

Case 1: Simple random sampling without replacement

Under the simple random sampling without replacement, we have that $\pi_i = \frac{n}{N}$ for $i \in U$, $\pi_{ij} = \frac{n(n-1)}{N(N-1)}$ for $i \neq j \in U$, $\pi_{ijk} = \frac{n(n-1)(n-2)}{N(N-1)(N-2)}$ for $i \neq j \neq k \in U$, and $\pi_{ijkl} = \frac{n(n-1)(n-2)(n-3)}{N(N-1)(N-2)(N-3)}$ for $i \neq j \neq k \neq l \in U$. It follows that

$$\pi_{ijk} - \pi_{ij}\pi_{k} = -\frac{2n(n-1)(N-n)}{N^{2}(N-1)(N-2)} = O(n^{-1}).$$

where the last equality is from Condition C.3. We also obtain

$$\begin{aligned} \pi_{ijkl} - 4\pi_{ijk}\pi_{l} + 6\pi_{ij}\pi_{k}\pi_{l} - 3\pi_{i}\pi_{j}\pi_{k}\pi_{l} \\ &= \left(\pi_{ijkl} - \pi_{ijk}\pi_{l}\right) - 3\left(\pi_{ijk}\pi_{l} - \pi_{ij}\pi_{k}\pi_{l}\right) + 3\left(\pi_{ij}\pi_{k}\pi_{l} - \pi_{i}\pi_{j}\pi_{k}\pi_{l}\right) \\ &= 3\frac{n(n-1)(n-2)(n-N)}{N^{2}(N-1)(N-2)(N-3)} - 6\frac{n^{2}(n-1)(n-N)}{N^{3}(N-1)(N-2)} + 3\frac{n^{3}(n-N)}{N^{4}(N-1)} \\ &= O(n^{-2}), \end{aligned}$$

where the last equality is from Condition C.3. Thus, Condition C.4 holds under the simple random sampling without replacement.

Case 2: Poisson sampling

From the independence of Poisson sampling, $\pi_{ij} = \pi_i \pi_j$ for $i \neq j \in U$, $\pi_{ijk} = \pi_i \pi_j \pi_k$ for $i \neq j \neq k \in U$, and $\pi_{ijkl} = \pi_i \pi_j \pi_k \pi_l$ for $i \neq j \neq k \neq l \in U$. Hence, $\pi_{ijk} - \pi_{ij} \pi_k = 0$, and $\pi_{ijkl} - 4\pi_{ijk} \pi_l + 6\pi_{ij} \pi_k \pi_l - 3\pi_i \pi_j \pi_k \pi_l = 0$. It follows that Poisson sampling satisfies Condition C.4.

A.6 A lemma for proving Theorem 4

Lemma 1. For the HT estimator \hat{t}_{HT} and the IHT estimator \hat{t}_{IHT} , under the Conditions C.1-C.4, we have

$$E\left(\hat{\overline{t}}_{HT}-\overline{t}\right)^4=O\left(n^{-2}\right), \quad and \quad E\left(\hat{\overline{t}}_{IHT}-\overline{t}\right)^4=O\left(n^{-2}\right).$$

Proof. Noting that

$$\widehat{t}_{\mathrm{HT}} - \overline{t} = rac{1}{N} \sum_{U} rac{I_k - \pi_k}{\pi_k} y_k \triangleq rac{1}{N} \sum_{U} J_k y_k,$$

we have

$$\begin{split} \left(\hat{t}_{\mathrm{HT}} - \overline{t}\right)^{4} &= \frac{1}{N^{4}} \sum_{k} \sum_{l} \sum_{i} \sum_{j} (J_{k} y_{k}) (J_{l} y_{l}) (J_{i} y_{i}) (J_{j} y_{j}) \\ &= \frac{1}{N^{4}} \sum_{U} (J_{k} y_{k})^{4} + \frac{4}{N^{4}} \sum_{k \neq l} (J_{k} y_{k})^{3} (J_{l} y_{l}) + \frac{3}{N^{4}} \sum_{k \neq l} (J_{k} y_{k})^{2} (J_{l} y_{l})^{2} \\ &+ \frac{6}{N^{4}} \sum_{i \neq k \neq l} (J_{i} y_{i})^{2} (J_{k} y_{k}) (J_{l} y_{l}) + \frac{1}{N^{4}} \sum_{i \neq j \neq k \neq l} (J_{i} y_{j}) (J_{j} y_{j}) (J_{k} y_{k}) (J_{l} y_{l}) \\ &\triangleq \mathrm{I} + \mathrm{II} + \mathrm{III} + \mathrm{IV} + \mathrm{V}. \end{split}$$

For the first term I, using $\lambda \leq \pi_k \leq 1$ and $|I_k - \pi_k| \leq 1$ for any $k \in U$, we get

$$|E(I)| = E\left(\frac{1}{N^4}\sum_{U} (J_k y_k)^4\right) = \frac{1}{N^4}\sum_{U} \left(\frac{y_k}{\pi_k}\right)^4 E(I_k - \pi_k)^4 \le \frac{1}{N^4}\sum_{U} \left(\frac{y_k}{\pi_k}\right)^4 = O(n^{-2}).$$

For the terms II and III, we have

$$\left| E(J_{k}^{3}J_{l}) \right| = \left| \frac{1}{\pi_{k}^{3}\pi_{l}} E\left[(I_{k} - \pi_{k})^{3} (I_{l} - \pi_{l}) \right] \right| \le \frac{1}{\pi_{k}^{3}\pi_{l}} E\left[|I_{k} - \pi_{k}|^{3} |I_{l} - \pi_{l}| \right] \le \frac{1}{\pi_{k}^{3}\pi_{l}} \le \frac{1}{\lambda^{4}},$$

and

$$E(J_{k}^{2}J_{l}^{2}) = \frac{1}{\pi_{k}^{2}\pi_{l}^{2}}E\Big[(I_{k}-\pi_{k})^{2}(I_{l}-\pi_{l})^{2}\Big] \leq \frac{1}{\pi_{k}^{2}\pi_{l}^{2}} \leq \frac{1}{\lambda^{4}}.$$

Thus, $|E(II)| = O(n^{-2})$ and $|E(III)| = O(n^{-2})$. For the fourth term IV, we have that

$$\begin{split} \left| E\left(J_{i}^{2}J_{k}J_{l}\right) \right| &= \frac{1}{\pi_{i}^{2}\pi_{k}\pi_{l}} \left| E\left[\left(I_{i}-\pi_{i}\right)^{2}\left(I_{k}-\pi_{k}\right)\left(I_{l}-\pi_{l}\right)\right] \right| \\ &= \frac{1}{\pi_{i}^{2}\pi_{k}\pi_{l}} \left| \left(1-2\pi_{i}\right)\left[\left(\pi_{ikl}-\pi_{ik}\pi_{l}\right)-\pi_{k}\left(\pi_{il}-\pi_{i}\pi_{l}\right)\right] + \pi_{i}^{2}\left(\pi_{kl}-\pi_{k}\pi_{l}\right) \right| \\ &= O\left(n^{-1}\right), \end{split}$$

where the last step is from Conditions C.1 and C.4. It implies that $|E(IV)| = O(n^{-2})$. For the last term V, we have that

$$\frac{1}{N^{4}} E\left(\sum_{\substack{i \neq j \neq k \neq l}} (J_{i} y_{i}) (J_{j} y_{j}) (J_{k} y_{k}) (J_{l} y_{l})\right) \\
= \frac{1}{N^{4}} \sum_{\substack{i \neq j \neq k \neq l}} \frac{\pi_{ijkl} - 4\pi_{ijk} \pi_{l} + 6\pi_{ij} \pi_{k} \pi_{l} - 3\pi_{i} \pi_{j} \pi_{k} \pi_{l}}{\pi_{i} \pi_{j} \pi_{k} \pi_{l}} y_{i} y_{j} y_{k} y_{l} \\
= O(n^{-2}),$$

where the last step is from Conditions C.1 and C.4. Thus, $E\left(\hat{t}_{HT} - \bar{t}\right)^4 = O(n^{-2})$ holds.

Next we shall prove $E\left(\hat{\overline{t}}_{\text{IHT}} - \overline{\overline{t}}\right)^4 = O(n^{-2})$. Noting that

$$\widehat{\overline{t}}_{\mathrm{IHT}} - \overline{\overline{t}} = \frac{1}{N} \sum_{U} \frac{\overline{I_k} - \overline{\pi_k^*}}{\overline{\pi_k^*}} y_k = \frac{1}{N} \sum_{U} \frac{\overline{I_k} - \overline{\pi_k}}{\overline{\pi_k^*}} y_k + \frac{1}{N} \sum_{U} \frac{\overline{\pi_k} - \overline{\pi_k^*}}{\overline{\pi_k^*}} y_k \triangleq A + \Delta,$$

we have

$$E\left(\hat{\bar{t}}_{\rm IHT} - \bar{t}\right)^{4} = E\left(A + \Delta\right)^{4} = E\left(A^{4}\right) + 4\Delta E\left(A^{3}\right) + 6\Delta^{2}E\left(A^{2}\right) + 4\Delta^{3}E\left(A\right) + \Delta^{4}.$$
 (A.7)

Similar as the proofs of the result $E(\hat{t}_{HT} - \bar{t})^4 = O(n^{-2})$, using $\lambda \le \pi_k^* \le 1$, it is easy to obtain

$$E(A^{4}) = E\left(\frac{1}{N}\sum_{U}\frac{I_{k}-\pi_{k}}{\pi_{k}^{*}}y_{k}\right)^{4} = O(n^{-2}).$$
(A.8)

From equation (A.8), we have that $E(A^2) = O(n^{-1})$ and $E(A^3) = O(n^{-3/2})$. Meanwhile E(A) = 0 and

$$\Delta = \frac{1}{N} \sum_{U} \frac{\pi_{k} - \pi_{k}^{*}}{\pi_{k}^{*}} y_{k} = \frac{K}{N} \left(\frac{1}{K} \sum_{U_{2}} \frac{\pi_{k} - \pi_{k}^{*}}{\pi_{k}^{*}} y_{k} \right) = O(n^{-1}).$$

Therefore, from equation (A.7), we prove that $E\left(\hat{\overline{t}}_{\text{IHT}} - \overline{\overline{t}}\right)^4 = O\left(n^{-2}\right)$.

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