

# Quantifying cost-effectiveness of subsurface strata exploration in excavation projects through geostatistics and spatial tessellation

Yat Fai Leung\*, Wenfei Liu, Yangxue Lei, Shu-Chien Hsu

*Department of Civil and Environmental Engineering, The Hong Kong Polytechnic University, Hung Hom, Hong Kong*

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## Abstract

A major source of uncertainty in civil engineering projects arises from the geological and geotechnical variability at the project site. This paper presents an approach to quantify such uncertainty, and to rationally incorporate them into estimates of cost contingency. Contrary to the conventional approaches that rely on individual expert's opinion or data from past projects, the proposed approach allows site-specific assessments of the intrinsic geotechnical variability through geostatistical techniques. Such uncertainty can be reduced through geotechnical investigation, and spatial tessellation techniques are proposed to facilitate determination of the optimal locations of new boreholes. Cost-effectiveness of the boreholes can be evaluated based on the corresponding reductions in geotechnical uncertainty and their influence on the budget. The approach is illustrated using a hypothetical excavation scenario, where the project costs are affected by uncertainty in the subsurface strata, particularly the rockhead level across the site. Under the specific site conditions, a Pareto frontier is developed to reveal the relationship between the number of boreholes to be drilled and potential savings in contingency budget. Through this approach, the study promotes better utilization of geotechnical information and rational assessments of project risks associated with their variability, which may lead to improved project planning and resource allocation.

*Keywords:* Excavation, Contingency budget, Geotechnical variability, Delaunay triangulation, Voronoi cells

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\*Corresponding author  
Email address: [andy.yf.leung@polyu.edu.hk](mailto:andy.yf.leung@polyu.edu.hk) (Yat Fai Leung)

## 1. Introduction

Uncertainty in geotechnical engineering is a well-known, yet inadequately understood topic in the civil engineering profession. Delay and cost overrun of many large-scale infrastructure projects have been attributed to ‘unforeseen’ and complex geological and geotechnical conditions. According to a survey of 28 construction projects in the United Kingdom [1], more than 40% of the geotechnical problems encountered during construction arise from uncertainties related to the subsurface strata and the geotechnical properties. To reduce such uncertainties, geotechnical investigation (e.g., rock and soil sampling and testing) can provide additional information about the ground conditions at the site. However, there is currently no quantitative approach to relate this to the level of uncertainty across the site, or to elucidate how the project risks may be reduced through the additional information. Consequently, practitioners often rely on their individual experience or intuition when planning the geotechnical investigation programme. The qualitative nature of this practice makes it difficult to assess the cost-effectiveness of the investigation, or its implications on the overall budget and delivery time of the construction project. The problem can be exacerbated in infrastructure mega-projects, where delay in one part of the project often triggers cascading effects to the entire development plan. From a management standpoint, a cost or time contingency is usually included in the project budget or programme, as a common approach to control the risks of delay and cost overruns due to unforeseen conditions. In fact, the contingency budget or the ‘float’ of a particular task should be decided according to the level of uncertainty associated with the task. For excavation projects, it is therefore beneficial to quantify the geotechnical uncertainty, which then allows rational planning and apportioning of the risks.

Traditionally, cost contingency is incorporated as a simple percentage addition onto the base (cost) estimate, considering specific project features, past experience and historical data [2]. For large and complex projects, more rational estimates may be obtained either through deterministic or probabilistic approaches [3]. Some of the common deterministic approaches include linear regression

26 models, artificial neural networks (ANN) for more complex problems, and Least Squares Support  
27 Vector Machine (LS-SVM) in price variation modeling for construction management. For example,  
28 Sonmez et al. [4] proposed a linear regression model to predict the bidding contingency amount  
29 for contractors, by focusing on the major influential factors of contingency decisions identified from  
30 previous projects, while Thal et al. [5] developed a multiple linear regression model for similar  
31 purposes. Although the regression method outperforms the practice of assigning an arbitrary per-  
32 centage, a linear relationship may not be able to best fit the available historical data [6]. Therefore,  
33 Artificial neural networks (ANN) were utilized to perform nonlinear regression for more complex  
34 problems. These include the work by [7], who built a back propagation general regression neural  
35 networks (GRNN) to determine the cost contingency and allocation strategies at the preliminary  
36 stage. Lhee et al. [8] further proposed a two-step ANN-based model adding contingency rate as an  
37 intermediate output variable. Meanwhile, Cheng et al. [9] established a hybrid system based on  
38 Least Squares Support Vector Machine (LS-SVM) for modelling construction price variations, which  
39 can be used for decision making in construction management. Although these previous studies have  
40 illustrated the potentials of deterministic approaches, a few major criticisms remain regarding their  
41 applications. These include the heavy influence by subjectivity of individual experts, deficiency  
42 in accurately quantifying the project risks, and the fact that some of these techniques work like a  
43 ‘black-box’ [10, 11].

44 Probabilistic approaches were advocated to tackle these deficiencies. For example, Khalafallah  
45 et al. [12] proposed the Bayesian Belief Network to quantify project risk and uncertainty level, which  
46 allows the determination of the appropriate contingency percentage for construction projects. This  
47 approach was further developed by Kim et al. [13] to assess the probability of construction project  
48 delays based on case studies in developing countries. Meanwhile, other researchers adopted the  
49 Monte Carlo Simulation (MCS) to quantify cost contingency at different risk levels [14]. Since  
50 the risk factors in construction projects often contain both ‘random’ and ‘fuzzy’ variables [15], a  
51 Fuzzy MCS framework was established by Sadeghi et al. [16] to evaluate both components in the

52 estimation of contingency range.

53 These previous approaches mainly rely on historical data or qualitative experts' opinion and  
54 experience [17], with little discussion on the intrinsic source of uncertainty. This paper attempts  
55 to quantify a major source of uncertainty in excavation projects, arising from the geotechnical and  
56 geological variability at the project site. An automated strategy is proposed to quantify geotechnical  
57 uncertainty, and to evaluate its changes with additional boreholes in the project site. It accounts  
58 for site-specific geologic features based on the available existing information, which may include  
59 irregularly-spaced boreholes revealing variations of subsurface strata in different directions. The  
60 quantitative approach enables optimization to be performed to determine the number and locations  
61 of sampling points that lead to the most cost-effective investigation programme, with respect to  
62 the impacts on time and costs of the tasks. The proposed approach will be demonstrated through  
63 the scenario of an excavation project, where the major uncertainty arises from the variations of  
64 rockhead level across the site. Such variations heavily influence the quantity of rock materials to  
65 be excavated, and hence the planning of project budget and delivery time.

## 66 **2. Methodology**

67 This study utilizes the geostatistical approach discussed by Liu et al. [18] and Liu and Leung  
68 [19] to quantify the geotechnical variability associated with subsurface strata. Meanwhile, spatial  
69 tessellation techniques are adopted for the derivation of optimal geotechnical sampling strategies.  
70 Their cost-effectiveness can be evaluated through the reductions of uncertainty, and the subsequent  
71 implications on the budget and time of the excavation project. The three individual components  
72 of the proposed approach are described in the following sections.

### 73 *2.1. Quantification of geotechnical variability*

74 Liu et al. [18] and Liu and Leung [19] presented the details and verification of an integrated  
75 framework established to characterize the spatial variability of geological profiles and geotechnical

76 properties. This will be described briefly herein as it forms the basis of the sampling strategy  
 77 proposed in this study. In general, the spatial variations of the subsurface strata ( $\mathbf{z}$ ) can be  
 78 represented by a linear mixed model consisting of a large-scale trend ( $\mathbf{X}\boldsymbol{\beta}$ ), and the residual effects  
 79 ( $\boldsymbol{\varepsilon}$ ) that describe the deviations of the actual values from the trend (Fig. 1):

$$\mathbf{z} = \mathbf{X}\boldsymbol{\beta} + \boldsymbol{\varepsilon} \quad (1)$$

80 where  $\mathbf{X}$  is a matrix containing information of the spatial coordinates of sampled points, and  $\boldsymbol{\beta}$   
 81 represent the trend coefficients.  $\boldsymbol{\varepsilon}$  is often observed to be spatially correlated, with greater variations  
 82 between components  $\varepsilon_i$  and  $\varepsilon_j$  associated with larger separation distances between locations  $i$  and  
 83  $j$ . Accordingly, the variance of  $\boldsymbol{\varepsilon}$  can be represented by:

$$\mathbf{V} = \sigma_e^2 \mathbf{R} + \sigma_n^2 \mathbf{I} = (\sigma_e^2 + \sigma_n^2) [s\mathbf{R} + (1-s)\mathbf{I}] \quad \text{where } 0 \leq s = \frac{\sigma_e^2}{\sigma_e^2 + \sigma_n^2} \leq 1 \quad (2)$$

84 where  $\mathbf{I}$  is the identity matrix;  $\sigma_n^2$  are the random natural effects (white noise effects) which are  
 85 independent of separation distances;  $\sigma_e^2$  are the smooth scale variations, or the component of total  
 86 variance that correlate with separation distance, and such autocorrelation is described by the matrix  
 87  $\mathbf{R}$ .  $s$  is referred to as the spatial dependence, and represents the proportion of  $\sigma_e^2$  within the total  
 88 variance. Individual components of  $\mathbf{R}$  ( $R_{ij}$ ) describe the correlations between  $\varepsilon_i$  and  $\varepsilon_j$ , and the  
 89 relationship between  $R$  and separation distance ( $h_{ij}$ ) can be modeled by different mathematical  
 90 functions, such as the exponential, Gaussian (squared exponential), or spherical function, all of  
 91 which involve a parameter  $\theta$  that defines the range of correlation. Alternatively, the scale of  
 92 fluctuation ( $\delta$ ) is another parameter used to define the extent of the correlation [20], and is often  
 93 taken as the separation distance where the autocorrelation  $R$  drops to the value of 0.05. The  
 94 parameters  $\theta$  and  $\delta$  are related to each other according to the adopted correlation function. For  
 95 example,  $\delta \approx \sqrt{\pi}\theta$  for the Gaussian function.

96 Site-specific characterization of the spatial features mainly involves determination of the trend

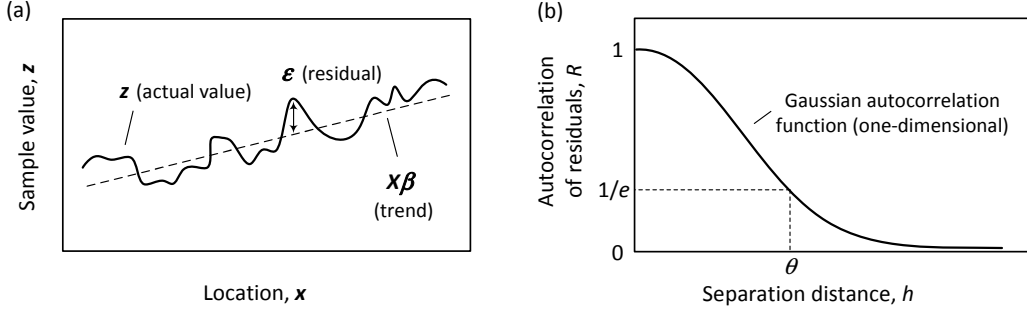


Figure 1: (a) Trend and residuals of spatial variables (after DeGroot and Baecher 1993); and (b) Autocorrelation of residuals

97  $\beta$ , together with correlation parameters  $s$  and  $\theta$ . These parameters can be grouped as a vector  
 98  $\Theta$ , and their values can be obtained through the ‘restricted maximum likelihood’ (REML) method  
 99 [21], which involves maximizing the following log-likelihood function:

$$L(\Theta|\mathbf{y}) = -\frac{n-m}{2} \log(2\pi) - \frac{1}{2} \log |\mathbf{V}| - \frac{1}{2} \log |\mathbf{W}| - \frac{1}{2} \mathbf{y}^T \mathbf{V}^{-1} \mathbf{Q} \mathbf{y}$$

$$\text{where } \mathbf{W} = \mathbf{X}^T \mathbf{V}^{-1} \mathbf{X}$$

$$\mathbf{Q} = \mathbf{I} - \mathbf{X} \mathbf{W}^{-1} \mathbf{X}^T \mathbf{V}^{-1} \quad (3)$$

100 In Eq. (3),  $n$  is the number of data points (e.g., geotechnical sampling points), and  $m$  is the  
 101 number of coefficients in the trend structure;  $\mathbf{y} = (\mathbf{I} - \mathbf{X}(\mathbf{X}^T \mathbf{X})^{-1} \mathbf{X}^T) \mathbf{z}$ , which is the vector of  
 102 filtered dataset with the trend components filtered out. The  $\mathbf{V}$  matrix depends on the correlation  
 103 structure (Eq. (2)), which in turn depend on the parameters in  $\Theta$ .

104 The determination of  $\Theta$  (that leads to maximum  $L(\Theta|\mathbf{y})$ ) can be treated as an optimiza-  
 105 tion problem, and various optimization techniques (e.g., gradient-based methods or evolutionary  
 106 algorithms such as genetic algorithm) can be applied for this purpose. Once  $\Theta$  is obtained,  
 107 predictions can be made for the ‘unknown’ properties at unsampled locations, through the ‘best  
 108 linear unbiased prediction’ approach. More importantly, the level of uncertainty can be quantified  
 109 through the prediction variance ( $\sigma_z^2$ ), which is based on the overall variance across the site and the  
 110 separation distances between the unsampled ( $\mathbf{x}_0$ ) and sampled ( $\mathbf{x}$ ) locations. Mathematically, this

111 is represented by:

$$\sigma_z^2 = \text{diag}(\mathbf{K}_0 - \mathbf{K}^T \mathbf{V}^{-1} \mathbf{K} + \mathbf{M}^T (\mathbf{X}^T \mathbf{V}^{-1} \mathbf{X})^{-1} \mathbf{M}) \quad (4)$$

112 where  $\mathbf{K}$  represents the covariance matrix between observations and predictions, i.e.,  $\mathbf{K} = \text{cov}\{\mathbf{z}(\mathbf{x}), \mathbf{z}(\mathbf{x}_0)\}$ ,  
 113  $\mathbf{K}_0 = \text{cov}\{\mathbf{z}(\mathbf{x}_0), \mathbf{z}(\mathbf{x}_0)^T\}$  and  $\mathbf{M} = \mathbf{X}_0^T - \mathbf{X}^T \mathbf{V}^{-1} \mathbf{K}$ , and  $\mathbf{X}_0$  is the deterministic component matrix  
 114 of prediction (unsampled) locations. Details about its derivation and implementation can be found  
 115 in Liu et al. [18], Liu and Leung [19], and Atkinson et al. [22].

116 It should be noted that the fundamental assumptions of the REML method require the data to be  
 117 stationary, which is sometimes shown (or assumed) to be the case in civil or geotechnical engineering  
 118 applications. Alternatively, the raw data can be transformed, e.g., through log transformation (to  
 119 become lognormally distributed dataset) or Box-Cox transformation [23, 18]. In these cases,  $\mathbf{z}$  and  
 120  $\boldsymbol{\varepsilon}$  in the above equations will represent the transformed data, and the evaluated  $\sigma_z^2$  will be under  
 121 the transformed space. These can be easily back-transformed to the corresponding values in the  
 122 original space.

## 123 2.2. Sampling strategies to minimize uncertainty

124 The main purpose of geotechnical investigation is to reduce the uncertainty associated with  
 125 the geological strata and geotechnical properties at the site. Its effectiveness can be evaluated  
 126 through the reduction in  $\sigma_z^2$  described by Eq. (4). Therefore, determination of the ‘best’ sampling  
 127 strategy can be considered as an optimization problem which, in theory, involves obtaining the  
 128 optimal configuration of boreholes (with coordinates  $\boldsymbol{\zeta}^*$ ) that minimizes  $\sigma_z^2$  (or the average of its  
 129 components) throughout the entire site domain:

$$\boldsymbol{\zeta}^* = \arg \min_{\boldsymbol{\zeta}} \frac{1}{N_\sigma} \sum_{i=1}^{N_\sigma} [\sigma_z(i)]^2 \quad (5)$$

130 where  $N_\sigma$  represents the number of locations across the site at which the prediction variance is  
 131 evaluated. A large-scale construction project will entail a large value of  $N_\sigma$  and also a large

132 matrix size for  $\mathbf{X}_0$ , which renders the optimization in Eq. (5) very time-consuming and sometimes  
 133 impractical. Therefore, this study adopts the Delaunay triangulation (DT) technique for efficient  
 134 optimization of the borehole configuration.

135 DT is often discussed together with the Voronoi diagram due to their dual properties, and the  
 136 formulations of the two are linked with each other. As shown in Fig. 2, for a domain that consists of  
 137  $n$  points (or ‘seeds’)  $p_i$  ( $i = 1, 2, \dots, n$ ), a set of  $n$  polyhedral convex regions ( $r_i$ , known as Voronoi  
 138 cells) can be formed. Within each Voronoi cell  $r_k$ , any point  $x$  is most strongly influenced by (or  
 139 located closest to) the seed of its cell  $p_k$ , compared to any other seeds. Mathematically, this is  
 140 represented by [24]:

$$r_k = \bigcap_{i \neq k} D(p_k, p_i) \quad i = 1, 2, \dots, n$$

where  $D(p_k, p_i) = \{x | h(p_k, x) \leq h(p_i, x)\}$  (6)

141 and  $h$  represents the separation distance between the two points. Once the Voronoi diagram is  
 142 constructed, the DT can be formulated by connecting the seeds that share a common edge in  
 143 their respective Voronoi cells. The DT possesses several important properties which make it very  
 144 versatile in computational geometry. One of these is that it maximizes the minimum angle of any  
 145 triangle [24, 25], which means the triangles formed by DT are the closest to equilateral triangles  
 146 compared to any other triangulation scheme.

147 In the current context, the boreholes in a geotechnical investigation programme are equivalent  
 148 to the ‘seeds’ of the domain, while the Voronoi cells, and edge lengths of the triangles, are associated  
 149 with the zones of influence of boreholes in different directions. Through the DT technique, sampling  
 150 configurations can be designed such that the influence of each borehole will be evenly spread across  
 151 the domain, thereby enhancing the overall effectiveness of the investigation programme. In other  
 152 words, when new boreholes are being planned at a site, the ‘optimal’ sampling strategy can be  
 153 derived by finding the configuration that minimizes the standard deviation of the edge lengths ( $t$ )



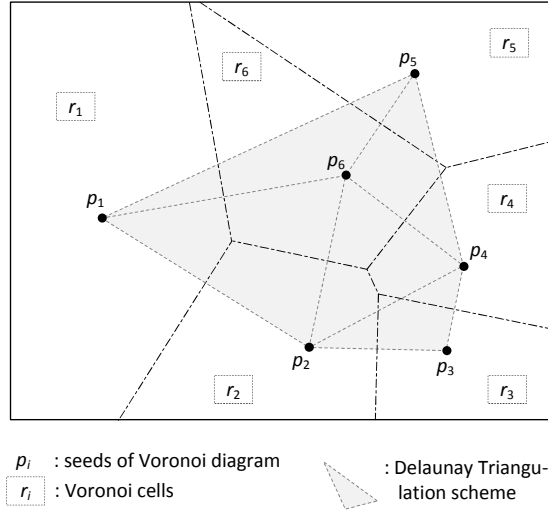


Figure 2: Voronoi diagram and Delaunay Triangulation

154 from DT, considering locations of both existing and new boreholes:

$$\zeta^* = \arg \min_{\zeta} \sqrt{\frac{1}{N_t - 1} \sum_{j=1}^{N_t} (t_j - \bar{t})^2} \quad (7)$$

155 which is an alternative way to Eq. (5) for optimal sampling, with  $N_t$  representing the total number  
 156 of edge lengths in the DT scheme, and  $\bar{t}$  being the average value of  $t$  across the domain. Due to its  
 157 widespread application, the DT has been programmed efficiently in many common software, making  
 158 it easier and faster to implement Eq. (7) than Eq (5). Once  $\zeta^*$  is determined, the corresponding  
 159  $\sigma_z^2$  can be evaluated by Eq. (4).

160 Fig. 3 demonstrates two simple cases where the optimal sampling strategies obtained through  
 161 the two different approaches are compared. The domain of the first case is 30 m  $\times$  30 m on plan,  
 162 with eight pre-existing boreholes at the corners and edges. The locations of three new boreholes  
 163 are to be determined, with the objective of minimizing the uncertainty in rockhead level ( $z$ ). The  
 164 second case involves a 30 m  $\times$  20 m domain, and one more pre-existing borehole is assigned near  
 165 the center, apart from the eight at the corners and edges. Five new boreholes will be planned  
 166 across the site for this case. It is further assumed that variance ( $\sigma_n^2 + \sigma_e^2$ ) for the residual of  $z$  is

167 100 m<sup>2</sup>, and  $s = 1$  for both cases. It is worth noting that the assumed variance corresponds to a  
168 standard deviation of 10 m in rockhead variations, while the assumption of  $s = 1$  is associated with  
169 the measurement error at sampled locations.  $s$  approaches 1 when the white noise effects ( $\sigma_n^2$ ) are  
170 small, and when there are no geologic faults causing abrupt changes in the rock level. The assumed  
171 values are close to the range revealed in another study by [18], and they do not actually affect the  
172 subsequent analysis results. This is because optimization of new borehole locations is controlled  
173 by the spatial distributions of  $\sigma_z^2$ , i.e., their changes with distances, that arise from the existing  
174 boreholes. The process will not be affected when  $\sigma_z^2$  is scaled up (or down) uniformly across the  
175 site by larger (or smaller) values of the variance. Meanwhile, the spatial correlation between the  
176 residuals at locations  $i$  and  $j$  is modeled by the Gaussian function:

$$R_{ij} = \exp\left(-\frac{h_{ij}^2}{\theta^2}\right) \quad (8)$$

177 where  $\theta$  equals 10 m for the first case and 5 m for the second. In addition, to simplify the scenario,  
178 a constant mean is adopted for the property so  $\mathbf{X}\boldsymbol{\beta}$  becomes a constant vector. This means the  
179 assumed trend of the rockhead is a horizontal plane. Again, the trend assumption does not influence  
180 the optimization process and the subsequent results, as the uncertainty arises mostly from the  
181 residuals, i.e. deviations of properties from the trend. When the framework is applied to actual  
182 field data, the trend is removed (or filtered out) to ensure stationarity conditions are satisfied.  
183 Subsequently, the residuals have zero mean; and the spatial correlation parameters and  $\sigma_z^2$  are all  
184 estimated based on this premise, and the trend coefficients  $\boldsymbol{\beta}$  are not involved in Eq. (4). By the  
185 same token, any trend assumptions may be adopted for the illustration cases, since  $\sigma_z^2$  is evaluated  
186 based on this same principle regardless of the trend.

187 Optimization of the additional borehole locations is performed by Differential Evolution, which  
188 is an evolutionary algorithm developed by Storn and Price [26] and recently adopted in a number  
189 of engineering applications. During the optimization process, the potential borehole configurations

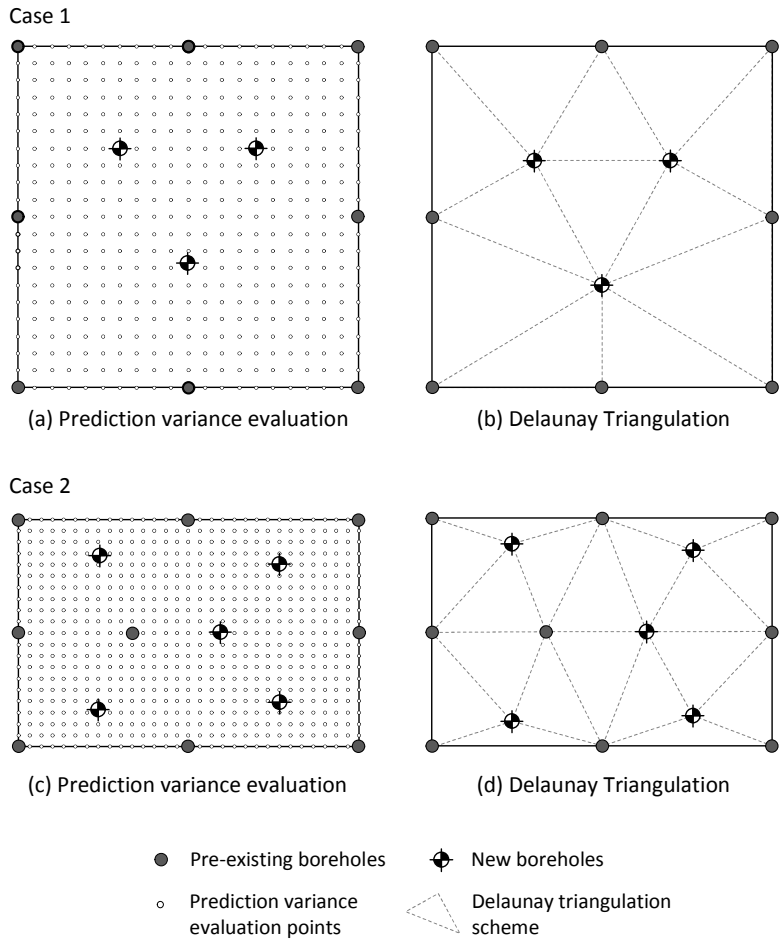


Figure 3: Optimal sampling locations for two illustration cases obtained by prediction variance evaluation and DT scheme

190 (candidate solutions) are expressed as vectors of variables (e.g., vectors of their coordinates) known  
191 as trial vectors. A population of trial vectors is first generated randomly, and the solution space  
192 is then explored through linear interpolation/extrapolation and mixing of trial vectors randomly  
193 selected from the population. Fitness of different trial vectors are evaluated and compared through  
194 an objective function, which determines the survivability of the particular solution: the fitter  
195 solutions stay in the population, while the weak ones will be discarded. The procedures are iterated  
196 until the population converges to a global optimum, and details of its implementation can be found  
197 in Storn and Price [26] or Leung et al. [27].

198 In the search for the optimal borehole configuration in this study, the objective function can  
199 be defined by Eqs. (5) or (7). In the former approach,  $\sigma_z^2$  needs to be evaluated for the entire  
200 domain, and discrete evaluation points are assigned with spacing of 1.5 m in both directions for the  
201 first case (i.e.,  $N_\sigma = 441$ ), and the spacing of these points is 1 m in the second case ( $N_\sigma = 651$ ).  
202 Figs. 3 shows that the optimal sampling strategies obtained by the two methods are very similar  
203 to each other, even when the two cases involve site domains of different sizes, shapes and pre-  
204 existing borehole arrangements. Despite very minor differences in the borehole arrangements, the  
205 corresponding discrepancies in the average  $\sigma_z^2$  are smaller than 3% between the two methods.  
206 Meanwhile, optimization with the DT scheme (Eq. (7)) is much faster since it is not necessary to  
207 evaluate  $\sigma_z^2$  for every potential configuration of new boreholes.

208 In this study, the locations of all additional boreholes are determined in a simultaneous manner.  
209 It is worth noting that multiple borehole locations may be determined also in a sequential manner:  
210 the first borehole is located where the uncertainty is largest, then the variance distribution is re-  
211 evaluated to determine location of the next borehole, and the process is repeated until reaching  
212 the desired total number of boreholes. The drawback of such approach, however, is that it does  
213 not guarantee optimum final arrangement of the multiple boreholes. Using the first case as an  
214 example, Fig. 4 compares borehole locations determined by the sequential and the simultaneous  
215 approaches, and the corresponding prediction variance contours. With the sequential approach, the

216 first borehole will be assigned at the center of the domain because this is the location furthest away  
217 from all existing boreholes, and corresponds to the largest uncertainty in the beginning. The second  
218 borehole will then be located midway between the center and one of the corners, with the third one  
219 close to another corner. The arrangements will be different from the optimal arrangement obtained  
220 by simultaneously considering three new boreholes (Figs. 3a and b). Figs. 4(e) and (f) also show  
221 that the sequential approach leads to a worse performance compared to the simultaneous approach:  
222 the average  $\sigma_z^2$  across the domain is about 7% higher with the sequential approach, which also  
223 gives rise to localized regions with high uncertainty, and the maximum value of prediction variance  
224 is 14% larger than that by the simultaneous approach.

### 225 *2.3. Cost-effectiveness of sampling strategies in excavation projects*

226 Quantity measurements in geotechnical engineering often depend on the subsurface geological  
227 profile at the site, and the variability of such would lead to uncertainties in the cost estimates.  
228 The cost-effectiveness of geotechnical sampling can be quantified through evaluating the associated  
229 reductions in uncertainty of the project costs. For example, for a building designed to be supported  
230 by end-bearing foundation piles, the actual cost of pile material and time of construction would  
231 depend on the depth of the competent firm stratum, on which the piles are to be founded. Increasing  
232 the number of boreholes would lead to better characterization of the stratum, reduced uncertainty  
233 and hence reductions of the contingency budget.

234 This study applies the concept to projects which involve excavation of soil and rock materials.  
235 Excavation of rock materials often costs three to five times more than that of soil materials, and the  
236 uncertainty associated with rockhead profile was cited as an important reason leading to the delay  
237 and cost overrun of a major underground railway station project in Hong Kong [28]. Following the  
238 earlier discussions on uncertainty of rockhead level ( $\sigma_z$ ), the total cost of excavation,  $C$ , can be

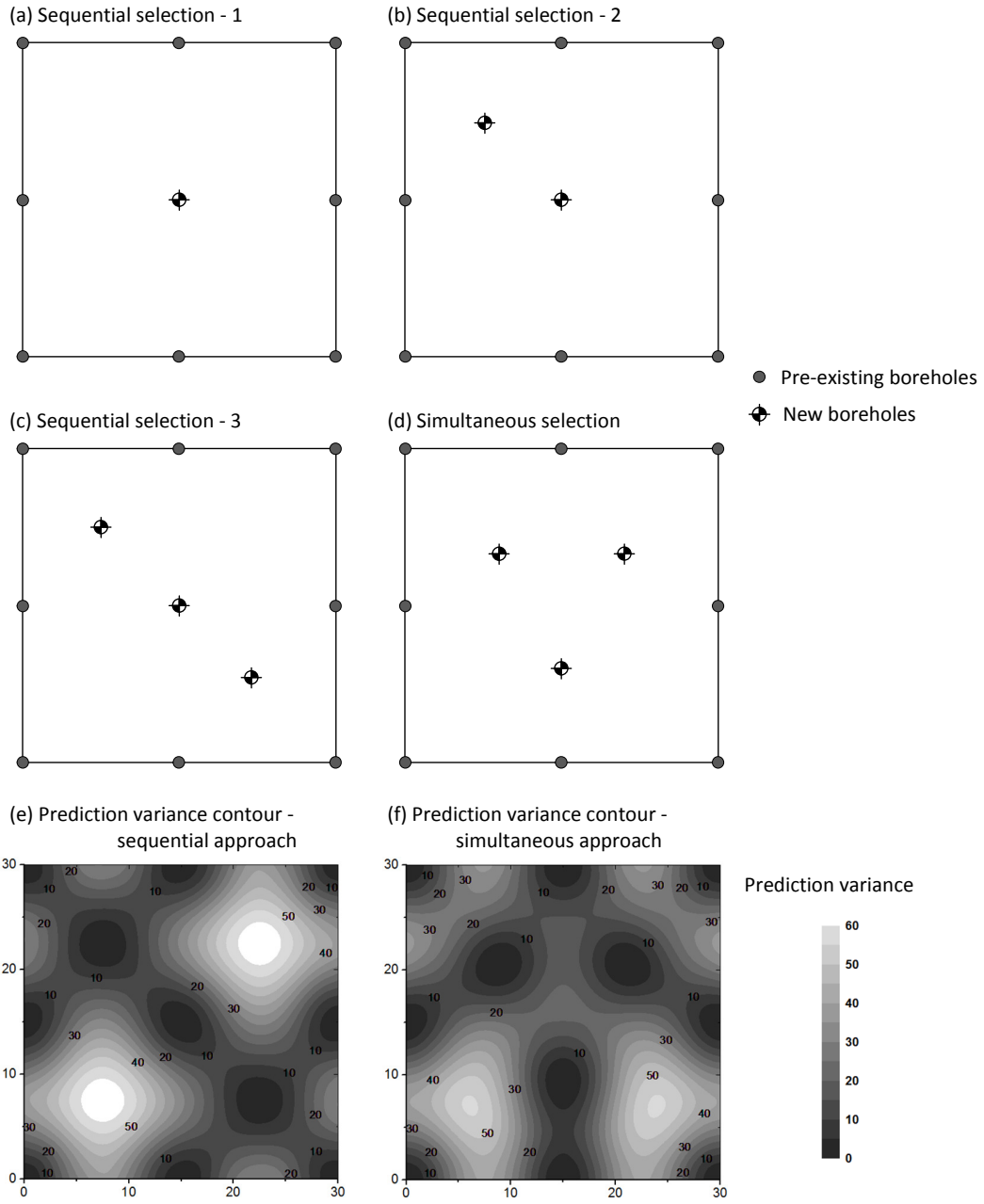


Figure 4: Comparisons between sequential and simultaneous determinations of new borehole locations

239 estimated for different ‘confidence levels’ of rockhead profile estimates:

$$C = \sum_{i=1}^{N_\sigma} A_i [C_r(d_i - d_{r,i}) + C_s(d_{r,i})]$$

where

$$d_{r,i} = \begin{cases} \hat{z}_i & \text{for 50\% non-exceedance probability} \\ \hat{z}_i - 0.675 \sigma_{z,i} & \text{for 75\% non-exceedance probability} \\ \hat{z}_i - 1.28 \sigma_{z,i} & \text{for 90\% non-exceedance probability} \end{cases} \quad (9)$$

240 and  $d_i$  represents the total depth of excavation (soil and rock) at location  $i$ ,  $A_i$  is the area associated  
 241 with that location,  $C_s$  and  $C_r$  are the unit rates of excavation in soil and rock, respectively;  $d_{r,i}$   
 242 is the depth of rockhead with different confidence levels:  $d_{r,i} = \hat{z}_i$  represents the best estimate at  
 243 location  $i$ , which can be obtained through simple interpolation or the best linear unbiased prediction  
 244 technique, while contingency in the budget can be introduced by subtracting from  $\hat{z}_i$  multiples of  
 245 the standard deviation of prediction at that location ( $\sigma_{z,i}$ ) (smaller  $d_{r,i}$  represents shallow rockhead,  
 246 which leads to larger volume of rock to be excavated).

247 The contingency budget,  $\tilde{C}$ , can also be expressed by rewriting Eq. (9) as follows:

$$\tilde{C} = \alpha \sum_{i=1}^{N_\sigma} A_i (C_r - C_s) \sigma_{z,i} \quad (10)$$

248 where  $\alpha = 0.675$  for 75% non-exceedance and 1.28 for 90% non-exceedance probabilities. In cases  
 249 where the raw data has been transformed (e.g., to the log-space) for REML analyses,  $d_{r,i}$ ,  $\hat{z}_i$  and  
 250  $\sigma_{z,i}$  in Eq. (9) and (10) should be represented in the ‘back-transformed’ original space. In the  
 251 current practice of project management, the valuation of risks involves much subjectivity. Apart  
 252 from providing a rational tool for quantification of uncertainty, Eq. (10) is proposed to allow risk  
 253 perception of different individuals to be incorporated through the parameter  $\alpha$ . Different values of  
 254  $\alpha$  correspond to different confidence levels, or in this context, different non-exceedance probability  
 255 for rockhead levels. For example, to cater for a high probability of non-exceedance (e.g., 90% chance  
 256 that the volume of rock excavation is not underestimated), a high value of  $\alpha$  (=1.28) is required and

257 that leads to a larger contingency budget. In cases where the project managers are more familiar  
258 with the site conditions, or where the risk tolerance is high, it is possible to adopt lower values of  
259 non-exceedance probability, and hence lower values of  $\alpha$  and contingency budget.

260 The sampling strategies described earlier aim to reduce  $\sigma_{z,i}$  through optimal configuration of  
261 additional boreholes, while Eqs. (9) and (10) quantify the effectiveness of drilling these boreholes  
262 with respect to the cost commitment of the project. These can be revised to also reflect the  
263 programme estimates for the excavation. As will be shown later, when the number of boreholes  
264 increases in the site, the return of investment associated with each additional borehole will gradually  
265 diminish. The quantification and optimization framework proposed in this study can help engineers  
266 make informed decisions regarding the optimal number of boreholes, as well as their best locations,  
267 for the specific project conditions.

### 268 **3. Illustration with hypothetical scenario**

269 The proposed approach is illustrated using a hypothetical excavation scenario. The information  
270 on subsurface geological profile, particularly the variations of rockhead level, is based on real data  
271 from a project site in Hong Kong, retrieved from the archive maintained by the Civil Engineering  
272 Library of the Hong Kong Government. Using the real data of geological variations, the site is  
273 hypothetically set up as an excavation project, to investigate the cost-effectiveness of uncertainty  
274 reduction by additional boreholes.

#### 275 *3.1. Site descriptions*

276 The data was obtained from a site in the Ngau Tau Kok (NTK) area in Hong Kong, and detailed  
277 geostatistical analyses using the REML method have been reported by Liu et al. [18]. The entire site  
278 domain is approximately 650 m  $\times$  450 m in area, with 150 irregularly-spaced, existing boreholes  
279 already drilled during previous development. The rockhead level is defined as the elevation of  
280 moderately decomposed granite, referred to as Grade III material according to GEO [29]. Based



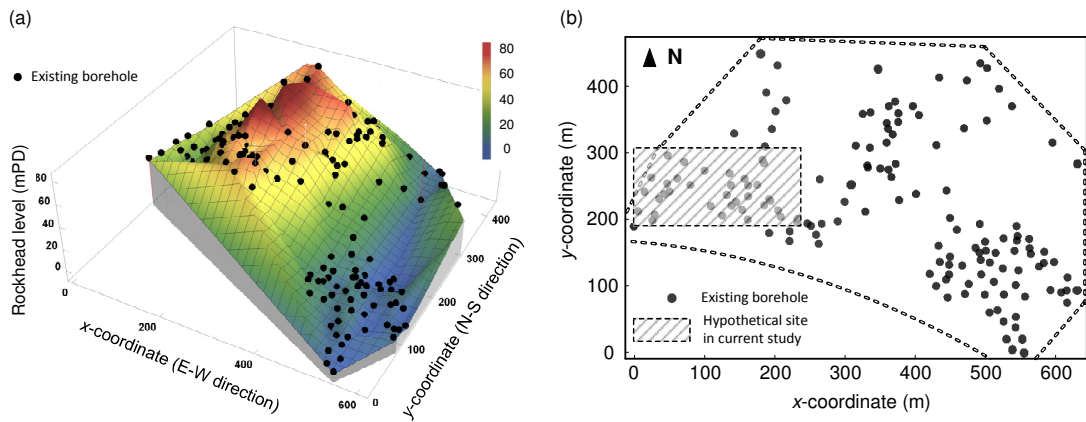


Figure 5: (a) Rockhead variations at NTK; (b) Study site within the NTK domain (modified from Liu et al. 2017)

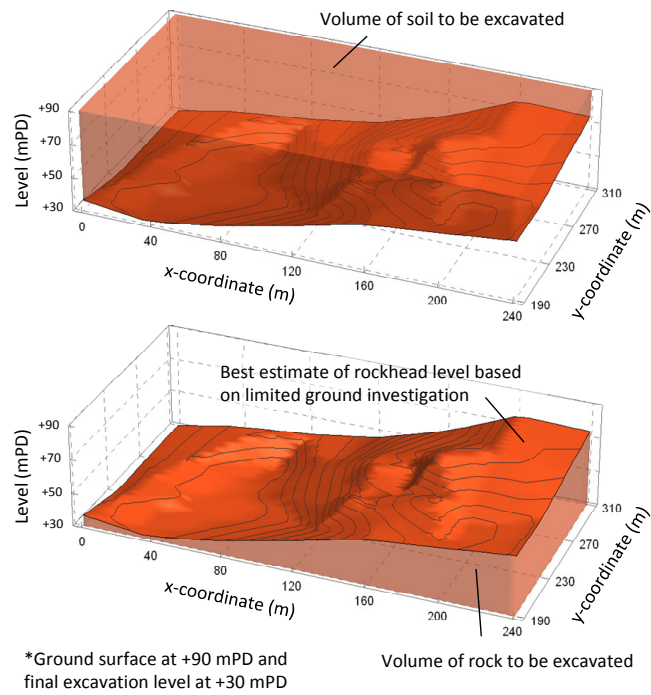


Figure 6: Volumes of soil and rock excavation at hypothetical site

281 on REML analyses (Eqs. (1) to (3)) of the existing data of rockhead levels, Liu et al. [18] reported  
282 the values of spatial dependence ( $s$ ) and scale of fluctuation ( $\delta$ ) to be 0.75 and 125 m, respectively.

283 In this study, a portion in the western side of the NTK domain is considered to be the site of a  
284 hypothetical excavation project. This hypothetical site is 240 m  $\times$  120 m in area (Fig. 5), where 35  
285 existing boreholes are located. It is further assumed that the ground surface and final excavation  
286 levels are at +90 mPD (Principal Datum) and +30 mPD, respectively. Therefore, the excavation  
287 depth is 60 m while the amount of rock and soil excavations vary (with uncertainty) across the  
288 site according to variations of the rockhead level. A graphical illustration of the project scenario is  
289 shown in Fig. 6. In the subsequent analyses, the unit rate of excavation is assumed to be HK\$150  
290 per m<sup>3</sup> in soil and HK\$500 per m<sup>3</sup> in rock (US\$1  $\approx$  HK\$7.8).

### 291 3.2. Determination of optimal sampling strategy

292 The uncertain rockhead levels across the site affect the cost estimates associated with excavation  
293 of soil and rock materials. The proposed approach is adopted with DT to obtain optimal configura-  
294 tions of additional boreholes to reduce such uncertainty. One major assumption associated with the  
295 following analyses is that the data from future additional boreholes will not change the correlation  
296 structure of rockhead levels (i.e.  $s$  and  $\delta$ ) obtained earlier. This assumption is inevitable during the  
297 planning stages of geotechnical investigation, since the data values at future boreholes, and hence  
298 their impacts on  $s$  and  $\delta$ , are unknown. In a real project scenario, it is possible to sequentially  
299 update  $s$  and  $\delta$  as more information becomes available during the exploration. This is, however,  
300 not performed for the hypothetical case in the current study.

301 Contrary to the scenario depicted in Fig. 3, the site boundaries in this hypothetical case only  
302 involve one existing borehole at the southwestern corner. Therefore, a number of auxiliary boundary  
303 points need to be assigned to define the boundaries for the triangulation scheme. To determine the  
304 spacing and locations of these auxiliary points, a DT analysis is first performed for the 35 existing  
305 boreholes (without considering the site boundary), and the median of edge lengths is found to be

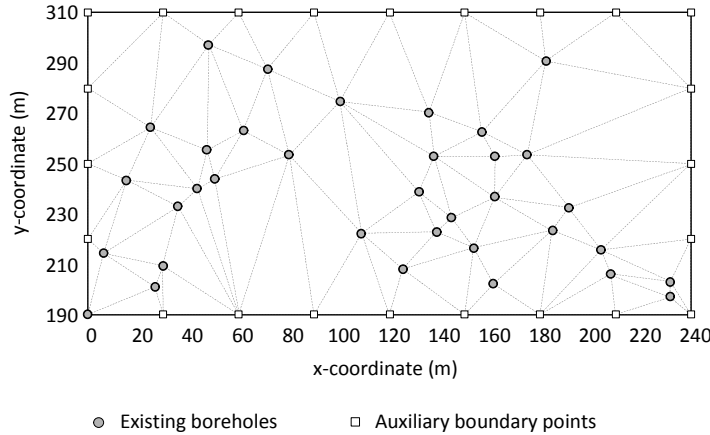


Figure 7: Delaunay triangulation scheme considering existing boreholes and auxiliary boundary points

306 approximately 30 m. The median is preferred over the mean since the latter appears to be skewed by  
 307 a few extreme values. This is then adopted as the spacing for auxiliary points along the boundaries  
 308 of the hypothetical site domain, as shown in Fig. 7. It should be noted that these auxiliary points  
 309 do not affect the estimates of prediction variance as they are not modeled as boreholes, and they  
 310 are also omitted in later figures for clarity. However, adopting different spacings for these points  
 311 will lead to slight influence on the results of triangulation, and hence the optimal sampling strategy.  
 312 These will be discussed further in a subsequent section.

313 In general, drilling more boreholes will always reduce the uncertainty, but that will also introduce  
 314 additional costs to the investigation programme. The cost-effectiveness of geotechnical investigation  
 315 at any particular site can be explored by developing the Pareto frontier, which reveals the rela-  
 316 tionship between the number of additional boreholes ( $n$ ) and the reduction in contingency budget.  
 317 For the current hypothetical site, this relationship is obtained through 30 individual optimization  
 318 analyses, each considering a different number of additional boreholes. The associated percentage  
 319 reduction in  $\tilde{C}$  is plotted in Fig. 8(a), for both 75% and 90% non-exceedance probabilities of  
 320 rockhead variations. For example, when 10 additional boreholes are drilled at the site, a reduction of  
 321 15% in  $\tilde{C}$  can be achieved, if  $\tilde{C}$  is estimated based on 90% non-exceedance probability of the amount

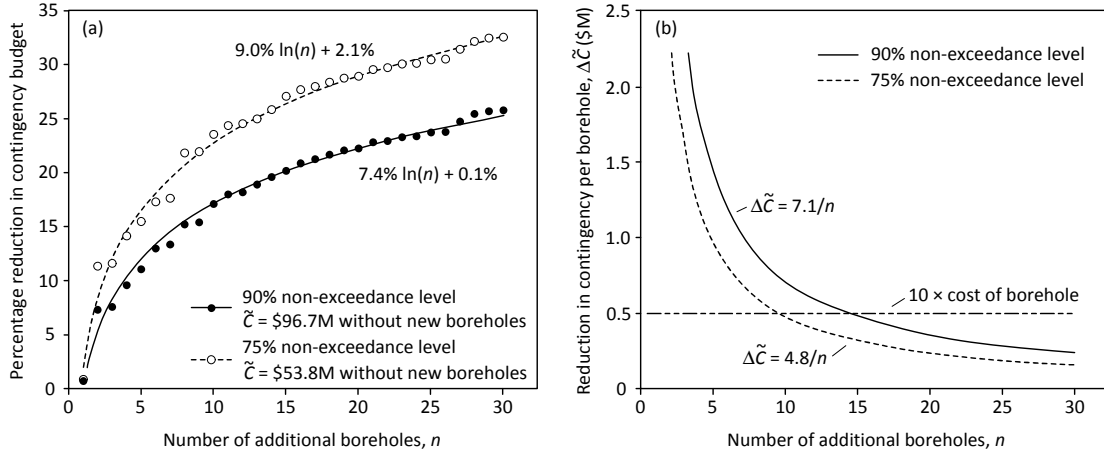


Figure 8: Pareto frontiers for 75% and 90% non-exceedance levels of rockhead variations

of rock materials to be excavated. The same number of boreholes can lead to 23% reduction in  $\tilde{C}$  estimated based on 75% non-exceedance probability. These percentage reductions are achievable when the layouts of additional boreholes are optimized using the proposed DT scheme. Under the current site settings, the 10 additional boreholes can lead to potential savings in contingency budget that exceed HK\$10 million.

The Pareto frontiers in Fig. 8(a) may be fitted by log curves also shown in the figure. The log relationships help to illustrate the effectiveness of the boreholes, as the slopes of the curves (which are proportional to  $1/n$ ) represent the extra values gained by adding one borehole to the investigation programme. This is denoted as  $\Delta\tilde{C}$  in Fig. 8(b). Although  $\Delta\tilde{C}$  drops with  $n$ , it far exceeds the drilling costs of a borehole under the adopted site settings. For example, assuming an approximate cost of HK\$1,250 per meter of drilling, a 40-m borehole would cost HK\$50,000. This is 10 times lower than the return of investment (through reducing  $\tilde{C}$ ) by the 15th borehole, based on 90% non-exceedance probability of rockhead level. Even after considering other subsidiary costs involved in sampling or mobilization of rigs, the return of investment is still substantial. The analyses also provide an indication of the balance point between costs and benefits of each additional borehole, and may be used as a quantitative illustration on the rationale behind the

338 optimum number of boreholes. As shown in Fig. 8(b), this optimum number also depends on the  
339 risk perceptions and risk tolerance at the project, reflected by the non-exceedance probability and  
340  $\alpha$  value (Eq. (10)) adopted in the decision-making process. It is important to note that under  
341 different project conditions, the associated drilling costs or contingency budget will differ from this  
342 hypothetical scenario, yet the proposed approach allows rational and quantified estimates to be  
343 obtained, and informed decisions to be made regarding the geotechnical investigation programme  
344 at the excavation site.

345 Fig. 9 presents two examples of optimal borehole configurations, with 10 and 20 additional  
346 boreholes, alongside the original layout of 35 existing boreholes. As mentioned earlier, the optimal  
347 arrangements are obtained by minimizing the standard deviations of edge lengths ( $t$ ) from the DT  
348 scheme. The distributions of  $t$  in the three configurations are also shown in Fig. 9. With more  
349 additional boreholes, both the average and standard deviations of  $t$  are reduced, which means the  
350 ‘coverage area’ for each borehole is decreased in a uniform manner, thereby effectively lowering  
351 the prediction variance or uncertainty across the site. These coverage areas can be illustrated  
352 graphically through the corresponding Voronoi cells, which become more uniform and smaller in  
353 size with increasing numbers of boreholes in an optimal arrangement.

354 The prediction variance contours for these borehole arrangements are shown in Fig. 10. It should  
355 be noted that although the color scale in the figures is capped at 50 for clarity, the average values  
356 around the edges of Fig. 10(a) (no additional boreholes) and Fig. 10(c) (20 additional boreholes)  
357 are approximately 200 and 60, respectively. The substantial reductions in prediction variance lead  
358 to the potential savings in contingency budget shown in Fig. 8.

#### 359 4. Discussions

360 In the previous section, it is assumed that the new data from additional boreholes do not affect  
361 the  $s$  and  $\delta$  parameters that characterize the spatial correlation structure. This is a reasonable  
362 assumption for the studied site since the parameters are obtained based on geostatistical analyses

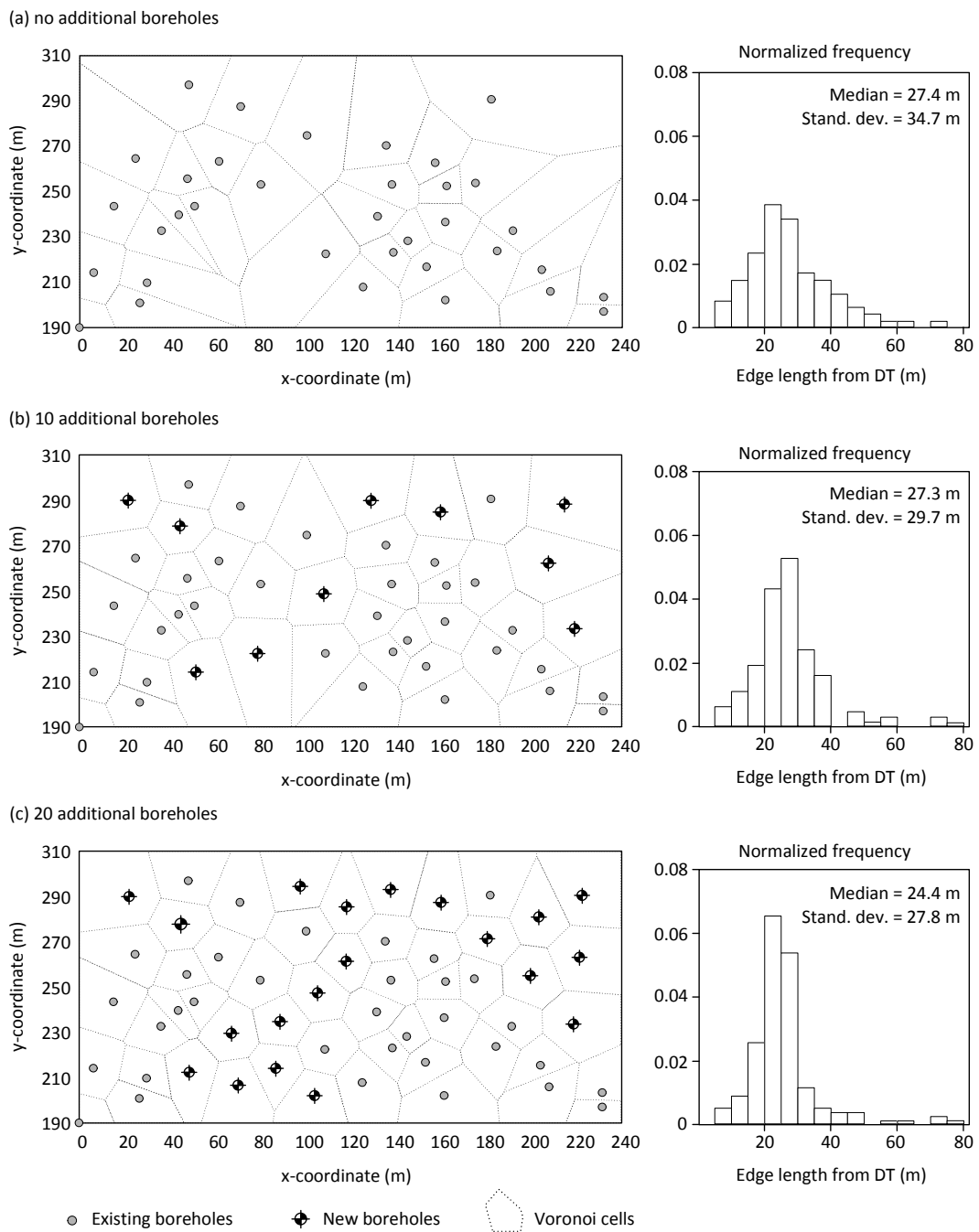


Figure 9: Optimal layout for (a) zero and (b) 10 and (c) 20 additional boreholes, with frequency distributions of edge lengths from Delaunay Triangulation

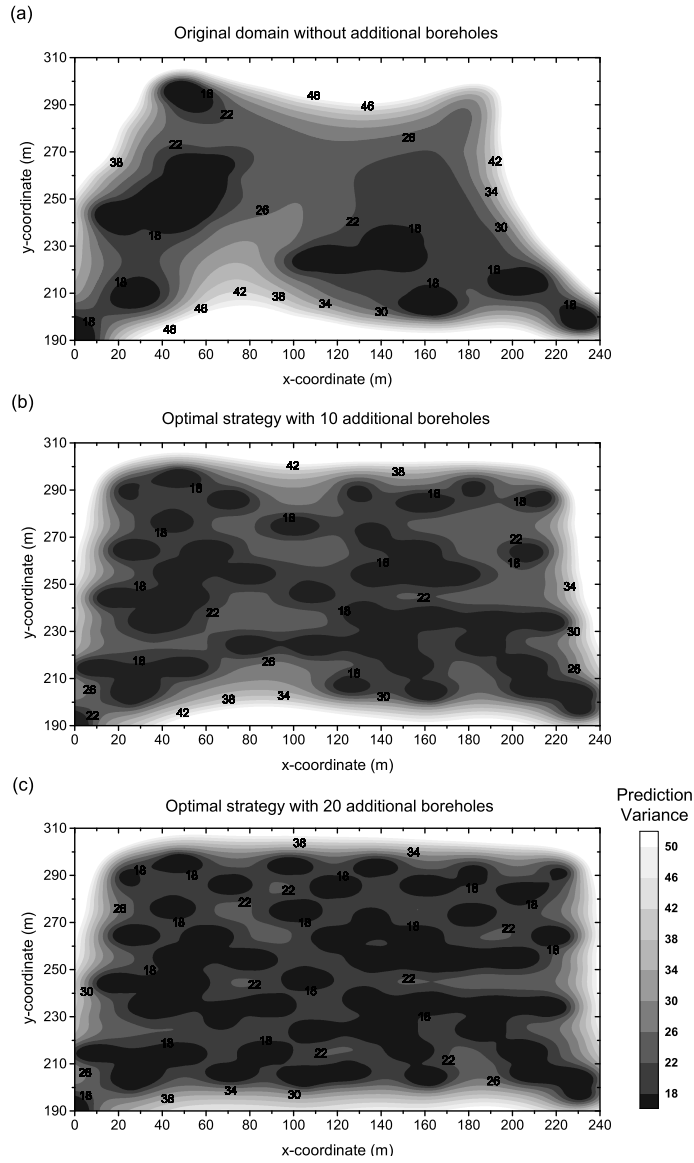


Figure 10: Prediction variance contours for (a) no additional boreholes; (b) 10 additional boreholes and (c) 20 additional boreholes

Table 1: Influence of auxiliary boundary points on Delaunay triangulation and prediction variance estimates

Spacing of boundary points (m)	10 additional boreholes			20 additional boreholes		
	20	30	40	20	30	40
Median of $t$ from DT (m)	26.8	27.3	27.7	24.7	24.4	25.2
Standard deviation of $t$ from DT (m)	29.5	29.7	29.0	25.9	27.8	23.8
Average prediction variance	38.7	37.4	37.6	33.9	31.6	29.7
Cost contingency (\$M)(90% non-exceedance)	80.5	80.2	79.8	76.1	75.2	73.5
(75% non-exceedance)	41.5	41.1	41.2	39.1	38.2	37.3

363 of 150 boreholes in the NTK domain, leading to relatively robust estimates. Similar analyses can  
364 be applied to other projects using information from previous developments around the region, or  
365 pre-tender investigation data. In cases where prior knowledge of the site is more limited, it would  
366 be advisable to progressively update the correlation parameters using newly available data, and  
367 reassess the cost-effectiveness of boreholes as the geotechnical investigation is being performed.  
368 Also, there may be situations where no prior information is available in the project area. In these  
369 cases, the first estimates of spatial correlation parameters may be reasonably assumed based on  
370 understanding of the local geology, or published information from the literature [18, 20, 30].

371 In preceding analyses of the hypothetical case, auxiliary boundary points are assigned along  
372 the site boundaries to facilitate the triangulation scheme, with the spacing of 30 m decided based  
373 on the median of edge lengths from DT of 35 existing boreholes. To explore the influence of the  
374 adopted spacing of these auxiliary points, additional analyses are performed, with the spacing set  
375 to be 20 m and 40 m, respectively. Table 1 compares these results with optimization of 10 and 20  
376 additional boreholes. In general, sparsely spaced auxiliary points lead to the additional boreholes  
377 being slightly closer to the boundaries of the domain. However, the overall impact on prediction  
378 variance, and hence the reductions in uncertainty or contingency, is not significant.

379 In the current study, the focus is placed on the impact of geological variability on the cost



380 uncertainty of the bulk excavation, i.e., removal of soil and rock materials. There are a number  
381 of other factors that influence the total costs of an excavation project, most notably the choice  
382 and installation of retaining wall and shoring systems. Although these factors are not evaluated  
383 explicitly in the current study, it is expected that they will also be influenced by data from additional  
384 boreholes. For example, reduced uncertainty in rockhead levels can lead to more accurate estimates  
385 on the depths of retaining walls. In fact, Eqs. (9) and (10) may be revised to assess the influence of  
386 reduced uncertainty on the potentials of project delays and the ensuing liquidated damages. The  
387 cost-effectiveness of geotechnical investigation would be even more significant if these benefits are  
388 also considered.

389 Apart from excavation projects, the proposed approach may be applicable to other civil and  
390 geotechnical project scenarios, where the quantity estimates and/or design assumptions are heavily  
391 influenced by uncertainty in subsurface strata. For example, the designs of both onshore and  
392 offshore foundation systems rely on proper understanding of the underground geological conditions  
393 and depths of firm strata supporting the foundations. The designs and cost estimates will benefit  
394 from rational decisions on the optimal planning of geotechnical investigation.

395 It should be noted that the current study adopts the isotropic autocovariance structure for  
396 variations in geological profiles, with the same spatial variability properties in both directions.  
397 This, however, may be inadequate when modelling certain geotechnical properties (e.g., shear  
398 strength and stiffness) that display anisotropic spatial correlation features in the three-dimensional  
399 subsurface domain. A potential extension of this study may focus on addressing this challenge in  
400 the future.

## 401 **5. Conclusion**

402 This study illustrates how the uncertainty in subsurface geological strata can be quantified  
403 through geostatistical analyses of the existing borehole information, and the benefits of incorpor-  
404 ating such considerations in the cost estimates of excavation projects. To maximize the cost-

405 effectiveness of geotechnical investigation, the locations of boreholes can be optimized to cater for  
406 the specific conditions of the project. This is achieved in the current study through coupling an  
407 optimization algorithm, the Differential Evolution, with spatial tessellation techniques including  
408 generation of Voronoi diagrams and Delaunay triangulation.

409 The proposed approach is demonstrated through analyses of a hypothetical excavation site  
410 consisting of real data of subsurface geological variations. It is shown that the contingency budget  
411 for excavation of soil and rock materials can be rationally and substantially reduced by considering  
412 the influence of additional boreholes located in optimal configurations. In addition, Pareto frontiers  
413 are developed to reveal the relationship between the number of additional boreholes and the cost-  
414 effectiveness of geotechnical investigation. In general, the benefits of geotechnical investigation  
415 to the finances and programme of the project will vary according to the specific site conditions.  
416 Nonetheless, the proposed approach can be a versatile tool applicable to various civil engineering  
417 projects, as it provides a platform where the assumptions and risks associated with subsurface  
418 profiles can be properly communicated, allowing various options of site investigation to be compared  
419 quantitatively in an evidence-based decision making process.

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