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RESEARCH ARTICLE

Sensitivity Analysis and 3D-displacement Inversion of Rock Parameters for High Steep Slope in Open-pit Mining

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Abstract: Due to the complexity of multiple rocks and multiple parameters circumstance, various parameters are often reduced to only one parameter empirically to generalize geological conditions, ignoring the really influential parameters. A developed method was presented as a complement to 3D displacement inversion to obtain the relative important parameters under complex conditions with limited computational work. Furthermore, this method was applied to a high steep slope in open-pit mining to investigate field applicability of the developed system. Back analysis was conducted in the reality of the east open-pit working area of Daye Iron Mine and propositional steps were presented for parameters solving in complex circumstance. Firstly, multi-factor and single-factor sensitivity analysis were carried out to classify rock mass and mechanical parameters respectively according to the extent of their effects on deformations. Secondly, based on the results, main influence factors were selected as inversion parameters and taken into a 3D calculating model to get the displacement field and stress field, all of which would be the artificial network training samples together with inversion parameters. Thirdly, taking the real deformations as input for the trained back propagation (BP) neural network, the real material mechanical parameters could be obtained. Finally, the results of trained neural network have been confirmed by field monitoring data and provide a reference to obtain the matter parameters in complicated environment for other similar projects.

Keywords: Back analysis, BP neural network, High steep slope, Rock mechanical parameters, Sensitivity analysis.

1. INTRODUCTION

The rock mass mechanical parameters are the key factors for studying the stability of slope in open-pit mines. Nowadays, the methods and techniques for open-pit slope stability analysis is multifarious, for instance using limit equilibrium methods, numerical methods, probabilistic approaches, *etc.* [1]. And more recent innovative methods, like Slope Quality Index (SQI) and Rock Engineering Systems (RES) approach incorporate a systemic procedure to estimate slope stability [2].

In the last decades, the numerical method has been applied extensively in current engineering practices to mitigate the complexity and uncertainty of geological factors for a long time. However, the validity in numerical methods is directly or indirectly related to the accuracy of rock mass mechanical parameters of calculations in most practical rock engineering projects.

One powerful approach to tackle this problem is parameter inversion with numerical simulation methods, of which the results were used as the equivalent rock mechanical parameters to reflect something unknown in the whole geotechnical system. Gioda [3] and Sun [4] explored fundamentals principles based on recent developments of the numerical techniques for parameter inversion. Tang *et al.* [5] published systematical treatise about engineering geology numeral simulation and back analysis of parameters. Levasseur *et al.* [6] set out the identification of constitutive

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parameters from *in-situ* geotechnical measurements. Kveldsvik [7] and Cai [8] back-calculated rock mass strength parameters based on measured displacements and geotechnical data. Li *et al.* [9] experienced how the equivalent parameters obtained and affected the estimates of safety factors.

Scholars presented practical new methods and obtained a practically unique solution for the elasto-plastic back analysis for rock mechanical parameters [10 - 12]. Artificial neural network (ANN) and genetic algorithm were also applied to parameters inversion to get the optimized solution of mechanical parameters in recent years [13 - 16], and typical examples were given in different circumstances to prove the feasibility and accuracy, such as powerhouse cavern [17, 18], large landslides [19], excavation support systems [20], sequential tunneling [21, 22], complex rock slope deformation [23 - 25].

The essential restriction of parameter inversion is that the parameters of compute should be established at the outset: if do not know the purpose of the study, we cannot determine which variables are relevant. So, when there are different types of rocks in the studying area, scholars often simplify geological conditions and select only one of those various parameters or formations according to their project experience. Yet amid to simplify calculations, there are concerns that the reductionist approach is poorly precise and even fail to serve its purpose. For mining excavation calculation in particular, big area coverage, long excavating cycle, there are very complex overlying strata and mechanical parameters of rock mass involved in calculation, the hypothesis of singular one media or excavating all at once cannot satisfy the needs of engineering.

Despite the existence of back-analysis, a more complete one that is able to combine a broader number of potential relevant factors and parameters of complex rock engineering projects is still lacking. Thereby, in this paper, a newly approach that integrates the features of 3 dimensional (3D) displacement inversion and sensitivity analysis was developed to infer the uncertainties under complex geological conditions from a holistic point of view and obtain the really matter sensitive parameters with finite calculations, and the east open-pit working area of Daye Iron Mine in China were selected as the case study examples.

2. METHODOLOGY

As far as a practical mining excavation calculation model, the complex nonlinear relationship between the characteristic factors of rocks and their mechanical behaviors by explicit functions may be unquantifiable. Once all the potentially relevant relations have to been taken into account, the given parameters of compute will be enormous and more difficult to determine. A new and simplified approach developed from 3D-displacement inversion is presented for the determination of rock mechanics parameters, to this system, multi-factor sensitivity analysis and mono-factor sensitivity analysis are added to improve it. Fig. (1) illustrates the flow chart of this study.

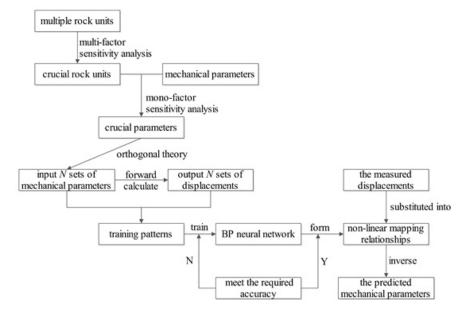


Fig. (1). The flow chart of the improved inversion method.

The major steps for obtaining rock parameters are organized as follows:

- 1. When the 3D numerical calculation model of study area have been established, all rock units in the area should be ranked firstly by multi-factor sensitivity analysis according to their influence on the deformation of 3D numerical model.
- Then sort the mechanical parameters of the relative important rocks in terms of vulnerability to ground deformation by mono-factor sensitivity analysis. After the relative importance of each rock units and mechanical parameters determined, the parameters with larger weights will be chosen as inversion parameters for computing.
- 3. Take the inversion parameters (*E*, μ , *C*, φ *et al*.) into 3D numerical model to calculate the displacement field and stress field (σ , u), both the two sets of data, the inversion parameters and numerical results (σ , u)), make up the training samples for artificial neural network (ANN).
- 4. When the training is completed from the training samples, the well-trained neural network could excellent map the nonlinear relation between parameters and numerical results. And then input the measured displacement (u^*) into the trained network as predicting sample to obtain the real material parameters (E^* , μ^* , C^* , φ^* *et al*).
- 5. Apply the real material parameters (E^* , μ^* , C^* , $\varphi^*et al.$) into 3D model, and the stress and the displacement deformation are obtained from the forward calculation results, which are verified by the field monitoring data to ensure the accuracy of parameters.

The results can meet the requirement of the project and provide a theoretical basis for long-term stability of super high steep slopes.

3. PRACTICAL APPLICATIONS

3.1. Sampling Location and 3D Model

Fig. (2) shows the aerial photo of Daye Iron Mine and the open-pit slope. The East Pit of Daye Iron Mine is 2.4 km long in *E-W* direction and 1.0 km long in *S-N* direction. The height of north slope of Daye open-pit is within 170~270 m and that of the south slope is within 86~200 m. The high-steep slope here is 230~430 m high and the slope angle varies between 38° and 43°. Open pit method is adopted. Fig. (3) shows the location of the selected study area in the East Pit of Daye Iron Mine.



Fig. (2). The aerial photo of Daye Iron Mine and the open-pit slope.

The study area's geological structure is medially developed. Most rock masses are jointed solid rock, and there are two main faults: F9 and F25, both of which pass through the upper bed and the lower bed respectively. The *in-situ* stress test results indicate that no remarkable tectonic stress appears.

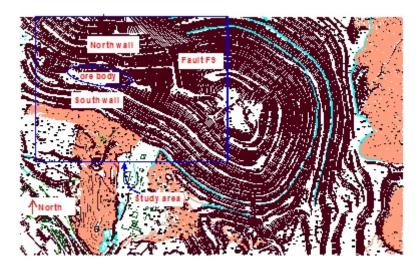


Fig. (3). Location of the selected study area (inside the blue rectangle) in the East Pit of Daye Iron Mine.

Mohr-Coulomb (M-C) model was considered as the material model for numerical simulation [26]. Considering the stress elimination, the model size's in the *E-W*, *N-S*, and vertical direction was 900 m, 800 m and 680 m. The study areas can be divided into five rock units: diorite, marble, magnetite, fault fracture zone and back filled body according to the type of rock masses. Tetrahedral meshes were used in the model, and composed of 76331 elements and 131016 nodes. Fig. (4) shows the 3D numerical calculation model.

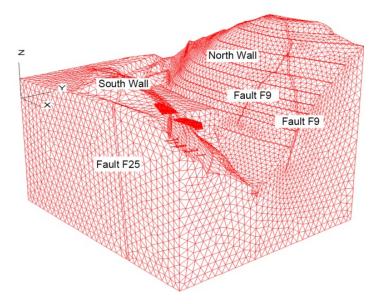


Fig. (4). Finite difference grid and 3D numerical calculation model.

In this calculation, five subsidence checkpoints were selected to process sensitively analysis. Checkpoints are represented by one of these letters U1, U2, U3, U4, U5, these letters correspond to locations south slope-diorite, north slope-marble, back filled body group, magnetite group and fault fracture zone, respectively.

3.2. Factors Definition with Sensitivity Analysis

According to the related investigation data and the preliminary research results [27], parameter values in Table **1** were regarded as the reference values for sensitivity analysis. 25 vast and complex different rock parameters need to be considered in the studying area. If every parameter of each rock units needed to be inverted, the number of combined schemes could be significantly large. Besides, output variables' number is too great to get each parameter accurately. For that, some measures, namely multi-factor and mono-factor sensitivity analysis, are taken to achieve less parameters and definite solution.

Rock units	Elasticity modulus /GPa	Poisson's ratio	Cohesion /MPa	Internal friction angle /(°)	Tensile strength /MPa
Marble	8.5	0.23	0.26	30	0.25
Diorite	9.0	0.23	0.27	29	0.10
Magnetite	7.8	0.27	0.35	42	0.35
Back filled body	3.0	0.22	0.07	27	0.02
Fault fracture zone	3.0	0.22	0.30	22	0.04

Table 1. Reference values for sensitivity analysis model.

3.2.1. Multi-factor Sensitivity Analysis of Rocks

Rocks should be firstly ranked by multi-factor sensitivity analysis according to their influence on deformation. In the part, factors were the five rock units, based on the reference values in Table 1, four levels (i=1, 2, 3, 4) were chosen: the reference value, reference value rise by 20%, reference value rise by 10%, reference value fall by 10%. Table 2 shows the 5-factor 4-level orthogonal experiment ($L_{16}(4^5)$) and the corresponding results of multi-factor sensitivity analysis.

Table 2. The orthogonal table and results.

No.	Factors and levels of orthogonal experiment								
	Marble	Diorite	Magnetite	Back filled body	Fault fracture zone				
1	1	1	1	1	1				
2	1	2	2	2	2				
3	1	3	3	3	3				
4	1	4	4	4	4				
5	2	1	2	3	4				
6	2	2	1	4	3				
7	2	3	4	1	2				
8	2	4	3	2	1				
9	3	1	3	4	2				
10	3	2	4	3	1				
11	3	3	1	2	4				
12	3	4	2	1	3				
13	4	1	4	2	3				
14	4	2	3	1	4				
15	4	3	2	4	1				
16	4	4	1	3	2				
S_1	330.5	153.5	265.7	291.9	310.5				
S_2	305.25	6.9	284.9	305.7	277.7				
S_3	283.35	34.3	313.7	270.65	294.55				
S_4	280.7	1005	335.3	331.45	317.4				
X_1	82.625	38.375	66.43	72.975	77.625				
X_2	76.312	1.725	71.22	76.425	69.425				
<i>X</i> ₃	70.837	8.575	78.42	67.662	73.637				
X ₄	70.175	251.25	83.83	82.862	79.35				
R	12.45	249.52	17.4	15.2	9.925				

 S_i (i=1,2,3,4): sum of four computational displacements of a certain level i;

 X_i (i=1,2,3,4): average value of the sum of four computational displacements, $X_i = S_i/4$;

R: the range of $X_p R = X_{max} - X_{min}$;

Levels (i=1, 2, 3, 4): the reference value, reference value rise by 20%, reference value rise by 10%, reference value fall by 10%.

For several factors simultaneously changed were considered. The range of the average value of factors (R) could reflect the differences of sensibility of the factors accurately. Analyzed the range 'R' of each factor in Table 2 and assessed affecting degree of each rock unit were diorite, magnetite, the back filled body, marble, and faults. These results were in coincidence with previous practical project experience.

3.2.2. Mono-factor Sensitivity Analysis of Parameters

The mono-factor sensitivity analysis was applied to rank the relative importance of mechanical parameters of each single type of rock mass. Firstly, 5 mechanical parameters of each rock mass (listed in Table 1) were taken as the fundamental state, $U = f(x_1, x_2, x_3, ...x_n)$. When the other entire parameters are fixed, the parameter x_i varied at a time with increasing by 20%, increasing by 10%, and reducing by 10%. If a small change of factor X in the inputs could produce the greatest effects on output U, then the displacements of monitoring points U was marked sensitive to the changes in factor x_i .

The overall parameters of the rock units are presented and for one specific rock mass, the diorite, the calculation of mono-factor sensitivity is presented in detail. The 5 mechanical parameters values (Table 1) were considered as base values and reasonably compounded in the calculation table of mono-factor sensitivity analysis to obtain the displacements U as shown in Table 3.

No.	Elasticity modulus /GPa	Poisson's ratio	Cohesion /MPa	Internal friction angle /(°)	Tensile strength /MPa	U ₁ /mm	U2 /mm	U ₃ /mm	U ₄ /mm
1	9.00	0.23	0.27	29.00	0.10	52.00	2.20	12.94	0.99
2	10.80	0.23	0.27	29.00	0.10	43.00	1.80	10.55	0.76
3	9.90	0.23	0.27	29.00	0.10	47.00	2.00	11.49	0.81
4	8.10	0.23	0.27	29.00	0.10	57.80	2.40	14.56	1.12
5									
6	9.00	0.28	0.27	29.00	0.10	48.00	2.85	13.18	1.77
7	9.00	0.25	0.27	29.00	0.10	52.00	2.50	13.73	1.40
8	9.00	0.21	0.27	29.00	0.10	51.00	1.96	14.28	0.60
9									
10	9.00	0.23	0.32	29.00	0.10	30.50	1.45	12.89	2.14
11	9.00	0.23	0.30	29.00	0.10	39.50	1.75	12.46	1.62
12	9.00	0.23	0.24	29.00	0.10	69.50	2.82	14.49	0.68
13									
14	9.00	0.23	0.27	34.80	0.10	9.20	1.33	14.25	4.13
15	9.00	0.23	0.27	31.90	0.10	24.00	1.60	12.93	2.65
16	9.00	0.23	0.27	26.10	0.10	158.00	3.50	16.24	1.70
17									
18	9.00	0.23	0.27	29.00	0.12	52.50	2.25	14.02	0.85
19	9.00	0.23	0.27	29.00	0.11	52.20	2.23	13.35	1.03
20	9.00	0.23	0.27	29.00	0.09	51.90	2.17	12.91	0.98

Table 3. Calculation table of mono-factor sensitivity analysis of diorite.

Experimental groups No.1, No.5, No.9, No.13, No.17 share the same value, namely the reference value of diorite.

 U_i is the displacements of four checkpoints computed by numerical calculation model. U_1 , U_2 , U_3 , U_4 , U_5 , correspond to locations south slope-diorite, north slope-marble, back filled body group, magnetite group and fault fracture zone, respectively.

For a quantitative comparison, the original combination indicators should be transformed into comparable values by the dimensionless method. Sensitivity of a normalized parameter was defined by the following formula [28]:

$$S_{k} = \left|\frac{\Delta U}{U}\right| / \left|\frac{\Delta x_{k}}{x_{k}}\right| = \left|\frac{\Delta U}{\Delta x_{k}}\right| \frac{x_{k}}{U} \quad k=1,2,\dots,n$$
(1)

where S_k is the sensitivity of factor x_k , $\left|\frac{\Delta U}{U}\right|$ is the relative variation of displacements; and $\left|\frac{\Delta x_k}{x_k}\right|$ is the relative variation of a given factor.

Plug mechanical parameters (Table 3) into the equation, then sensitivities of diorite's factors were calculated as the 'U₁' in Table 4. The sensitivity coefficients stand for the relative importance of mechanical parameters of diorite. Three factors were separated from the five mechanical parameters: internal friction angle (9.9615), cohesion (2.6122), and elasticity modulus (0.9817).

Monitoring points	Elasticity modulus /GPa	Poisson's ratio	Cohesion /MPa	Internal friction angle /(°)	Tensile strength /MPa
U_1	0.9807	0.2115	2.6122	9.9615	0.0353
U_2	0.9091	1.3106	2.1894	3.5379	0.1288
U_3	1.0995	0.5796	0.245	1.0211	0.5059
U_4	1.0746	4.0122	5.1041	19.6870	0.4133

Table 4. Sensitivity coefficients of each position's mechanical parameters.

 U_1 : checkpoint allocated in south slope-diorite; U_2 : checkpoint allocated in north slope-marble; U_3 : checkpoint allocated in back filled body group; U_4 : checkpoint allocated in magnetite group.

Similar to the exception of diorite, sensitivity coefficients of all the five monitoring positions' mechanical parameters were obtained as listed in Table 4. The sampling schemes for inversion are, in fact the parameter combinations of finite element calculation, and elasticity modulus are mandatory. According to the sensitivity coefficients, 11 parameters that have relatively larger impact on slope deformations were showed in descending orders are as follows:

U₁: Diorite in north slope: internal friction angle, cohesion, elasticity modulus;

U₂: Marble in south slope: internal friction angle, cohesion, elasticity modulus;

U₃: The back filled body: elasticity modulus, internal friction angle ;

U4: Magnetite: internal friction angle, cohesion, elasticity modulus.

With a view to the specificity of fault fracture zone, the following parameters were also chosen as the target vectors:

Faults: internal friction angle, cohesion, elasticity modulus.

Consequently there are a total of 14 pivotal parameters selected from 25 mechanical parameters to be calculated.

Table 5. Orthogonal design table for 14 levels of factor 2.

N								Leve	els						
No.	X1	X2	X3	X4	X5	X6	X7	X8	X9	X10	X11	X12	X13	X14	X15
1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	
2	1	1	1	1	1	1	1	2	2	2	2	2	2	2	
3	1	1	1	2	2	2	2	1	1	1	1	2	2	2	
4	1	1	1	2	2	2	2	2	2	2	2	1	1	1	
5	1	2	2	1	1	2	2	1	1	2	2	1	1	2	
6	1	2	2	1	1	2	2	2	2	1	1	2	2	1	
7	1	2	2	2	2	1	1	1	1	2	2	2	2	1	
8	1	2	2	2	2	1	1	2	2	1	1	1	1	2	
9	2	1	2	1	2	1	2	1	2	1	2	1	2	1	
10	2	1	2	1	2	1	2	2	1	2	1	2	1	2	
11	2	1	2	2	1	2	1	1	2	1	2	2	1	2	
12	2	1	2	2	1	2	1	2	1	2	1	1	2	1	
13	2	2	1	1	2	2	1	1	2	2	1	1	2	2	
14	2	2	1	1	2	2	1	2	1	1	2	2	1	1	
15	2	2	1	2	1	1	2	1	2	2	1	2	1	1	
16	2	2	1	2	1	1	2	2	1	1	2	1	2	2	

X1: elasticity modulus of diorite; X2: cohesion of diorite; X3: internal friction angle of diorite;

X4: elasticity modulus of marble; X5: cohesion of marble; X6: internal friction angle of marble;

X7: elasticity modulus of magnetite; X8: cohesion of magnetite; X9: internal friction angle of magnetite; X10: elasticity modulus of the fault fracture zone; X11: cohesion of the fault fracture zone; X12: internal friction angle of the fault fracture zone; X13: elasticity modulus of the back filled body; X14: internal friction angle of the back filled body; X15: null.

Level 1 and 2: represent reference value rise by 10% and fall by 10%.

3.3. Mechanical Parameters Inversion with Neural Network

3.3.1. Training Samples of Neural Network

The 14 acquired key parameters and a "null" were taken as 15 factors, and each factor was divided into 2 value

levels, namely each factor represent 110% and 90% of the baseline value respectively. Then, a orthogonal experiment design ($L_{16}(2^{15})$) was applied (Table 5), of which the combined projects { X_{1i} , X_{2i} , X_{3i} , ... X_{15i} }, i = 16 were regard as output vector.

Numerical tests of these combined projects were carried out with the 3D model (Fig. 4), the displacements of the five representative positions in different situations were calculated (listed in Table 6), and regard as the input vector.

Table 6. Displacement deformation of analysis model.

No.	U ₁	U ₂	U ₃	U_4	U ₅
1	-1.590	-2.508	0.010	0.068	0.248
2	-2.469	-3.308	0.040	0.032	0.276
3	-2.842	-4.002	0.000	0.045	0.257
4	-2.634	-3.646	-0.010	0.060	0.169
5	-8.311	-9.802	-0.018	0.021	0.195
6	-89.180	-105.100	-0.025	-0.048	0.201
7	-88.410	-104.800	-0.014	0.028	0.294
8	-89.750	-104.700	-0.032	0.022	0.185
9	-42.930	-49.210	-0.014	0.030	0.231
10	-41.870	-48.250	-0.014	0.044	0.234
11	-44.970	-52.140	-0.017	0.042	0.228
12	-43.240	-49.610	0.002	0.020	0.234
13	-8.293	-9.799	0.013	0.067	0.372
14	-9.402	-11.210	0.058	0.070	0.267
15	-8.628	-10.280	0.050	0.075	0.269
16	-8.347	-9.874	0.070	0.054	0.580

 U_1 : checkpoint allocated in south slope; U_2 : checkpoint allocated in north slope;

 U_3 : checkpoint allocated in back filled body group; U_4 : checkpoint allocated in magnetite group.

 $\mathrm{U}_{\scriptscriptstyle{5}}\!\!:$ checkpoint allocated in fault fracture zone.

By now, the training samples for neural network were acquired completely.

3.3.2. Neural Network Training

(1) Artificial neural network (ANN) design

The ANN model used here is based on the standard *BP* network architecture, which consists of fully interconnected layers of processing units, incorporated with transfer functions [29].

a. Input and output layer

The number of input neurons and output neurons depended on the number of elements in BP network. Input neurons were the displacements on 5 representative points and output neurons were 14 mechanical parameters, so a 5-input, 14-output network was necessary.

b. Hidden layer

A network with multiple hidden neurons can approximate any nonlinear multidimensional function. To improve the computation efficiency, one hidden layer was selected to form the network. Deciding the number of neurons in the hidden layers was a very important part of deciding the overall neural network architecture. Using too few neurons in the hidden layers will result in something called under-fitting. Under-fitting can lead to no convergence. Using too many neurons in the hidden layers can result in over-fitting. An inordinately large number of neurons in the hidden layers can also increase the time it takes to train the network. The amount of training time can increase to the point that it is impossible to adequately train the neural network. Obviously, some compromise must be reached between too many and too few neurons in the hidden layers. Generally, the number of hidden neurons can be decided in following formula [30]:

$$n_1 = \sqrt{n + m} + a \tag{2}$$

$$n_1 \ge \log_2 n \tag{3}$$

where n_1 is the number of hidden neurons; n is the number of output neurons, and n=5; m is the number of input

neurons, and m=14; *a* is a constant ranging from 1 to 10. Calculations indicate that the number of hidden neurons ranges between 5 and 15.

c. The modified BP algorithm

The Neural Network Toolbox provided tools for designing, implementing, visualizing, and simulating neural networks. Existing functions in MATLAB were called to implement the *BP* algorithms and the modified *BP* algorithms. A log-sigmoid transfer function (Logsig) was used to calculate the hidden layer's output, and a linear transfer function (Purelin) was employed to calculate the output layer's output.

(2) Training the network

Based on the relative research achievements and engineering experiences, $n_1=6$, $n_1=9$ and $n_1=12$ were selected as the numbers of hidden neurons. The data listed in Tables 5 and 6 were imported into the network as training samples. In the training process, the network prediction error went down incrementally until it fell below 10^{-4} , Fig. (5) shows the diagram of the considered *BP* network in the learning stage.

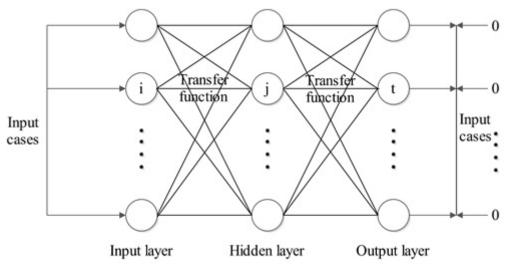


Fig. (5). Diagram of the considered BP network in the learning stage.

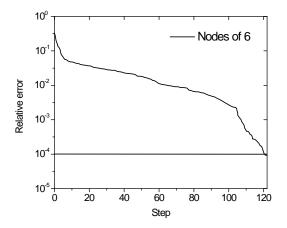


Fig. (6). Error curve for 6 nodes network structure.

Figs. (6-8) show the errors decrease with increasing epochs when $n_1=6$, $n_1=9$ and $n_1=12$ respectively. Results showed that the network with 9 hidden neurons performed better than others and could meet the requirement of accuracy.

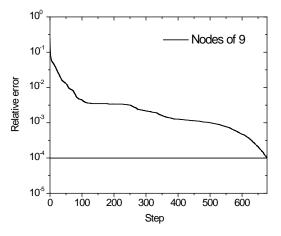


Fig. (7). Error curve for 9 nodes network structure.

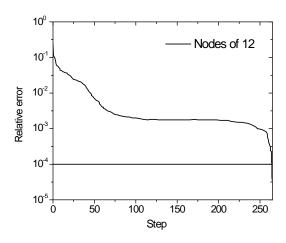


Fig. (8). Error curve for 12 nodes network structure.

3.3.3. Inverse Analysis of Mechanical Parameters

Once the ANN has been trained using the training set, the interaction function can be established with the non-linear function between mechanical parameters and displacements. The actual displacements of five monitoring points: U= [-35, -163, 0.75, -3.65, 1.12] (units: mm) was imported as the given input for the network with 9 hidden neurons. Then the output vectors would be the predicted mechanical parameters, as listed in Table 7.

Table 7. Rock mechanics parameters determined by back analysis.

Petrofabrics	Modulus of elasticity /GPa	Poisson's ratio	Cohesion /MPa	Internal friction angle /(°)	Tensile strength /MPa
Marble	15.60	0.23	0.25	28	0.25
Diorite	16.90	0.23	0.26	30	0.10
Magnetite	12.50	0.27	0.35	40	0.35
Back filled body	2.00	0.20	0.07	27	0.02
Fault fracture zone	1.70	0.20	0.13	23	0.04

3.4. Field Test Verification

The mechanical parameters calculated by inversion have great practical value in the surrounding rock stability and the slope stability of open-pit. The obtained parameters in Table 7 were applied to simulate open-pit excavation. Monitoring points F9-5 and F9-6 are located on both sides of Fault F9. Both the calculated vertical and horizontal displacements were compared with the field monitoring data to verify the the accuracy of the network. As listed in Table 8, the maximum error rate between numerically simulated data and field monitoring data was 13%, and the minimum error rate was as low as 0.2%. The basic agreement of those two sets of data indicates that the mechanical

parameters and initial stress field obtained from the proposed back analysis are reasonable.

No.		Observational data /mm	Computational results /mm	Relative error
F9-6	Z-Displacement	35.0	34.5	1.4%
Г9-0	X-Displacement	27.0	30.7	13.0%
E0.5	Z-Displacement	149.0	143.1	3.9%
F9-5	X-Displacement	112.0	112.2	0.2%

Verification analysis shows good agreements between the simulation results and the actual field data for slope stability in open-pit mines. As discussed in relation to the project example, this developed method could cull the pivotal parameters from amount of parameters correctly and effectively, and has good field applicability, not only for slope stability, but other rock engineering projects with complicated geological conditions.

CONCLUSION

- Both multi-factor sensitivity analysis and mono-factor sensitivity analysis were employed to determine the influence of rock types and mechanical parameters. In descending order, the lists of material types according their influence on slope deformations were diorite, magnetite, the back filled body, marble and the faults, which in coincidence with previous practical project experience. In descending order, the lists of mechanical parameters according their influence on slope deformations were internal friction angle, cohesion, elastic modulus, poisson ratio and tensile strength.
- 2. Take the obtained parameters into the 3D calculation model, the results showed that the small relative errors between numerical simulation result and the observation data meet the requirement of precision. So the 3D-displacement inversion with sensitivity analysis had great value in engineering applications of multi-parameters and provided a theoretical basis for long-term stability of super high steep slopes.
- 3. The method presented in this paper is a natural development of 3D displacement inversion, rock mechanical parameters could be successfully predicted and reliable for further study with finite steps. Improved by both multi-factor sensitivity analysis and mono-factor sensitivity analysis, this method could screen out the most efficacious rock mechanical parameters correctly and reasonably in complex environment with simple calculations. Furthermore, 5 reference detailed calculating steps were provided for parameter inversion in multiple rocks and multiple parameters circumstance.

CONFLICT OF INTEREST

The authors confirm that this article content has no conflict of interest.

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REFERENCES

- M. Pinheiro, S. Sanches, T. Miranda, A. Neves, J. Tinoco, A. Ferreira, and A.G. Correia, "A new empirical system for rock slope stability analysis in exploitation stage", *Int. J. Rock Mech. Min. Sci.*, vol. 76, pp. 182-191, 2015. [http://dx.doi.org/10.1016/j.ijrmms.2015.03.015]
- [2] M.Z. Naghadehi, R. Jimenez, R. KhaloKakaie, and S.E. Jalali, "A new open-pit mine slope instability index defined using the improved rock engineering systems approach", *Int. J. Rock Mech. Min. Sci.*, vol. 61, pp. 1-14, 2013. [http://dx.doi.org/10.1016/j.ijrmms.2013.01.012]
- G. Gioda, and S. Sakurai, "Back analysis procedures for the interpretation of field measurements in geomechanics", Int. J. Numer. Anal. Methods Geomech., vol. 11, pp. 555-583, 1987.
 [http://dx.doi.org/10.1002/nag.1610110604]
- [4] J. Sun, and W. Huang, "An Optimization method for the elastoplastic inversion of parameters in rock mechanics", Chin. J. Rock Mech. Eng., vol. 11, pp. 221-229, 1992.

- [5] H.M. Tang, E.C. Yan, and X.L. Hu, Principles and Applications of Numerical Modeling in Engineering Geology., China University of Geosciences Press: Wuhan China, 2002.
- S. Levasseur, Y. Malécot, M. Boulon, and E. Flavigny, "Soil parameter identification using a genetic algorithm", *Int. J. Numer. Anal. Methods Geomech.*, vol. 32, pp. 189-213, 2008.
 [http://dx.doi.org/10.1002/nag.614]
- [7] V. Kveldsvik, H.H. Einstein, B. Nilsen, and L.H. Blikra, "Numerical analysis of the 650,000 m2 Åknes rock slope based on measured displacements and geotechnical data", *Rock Mech. Rock Eng.*, vol. 42, pp. 689-728, 2009. [http://dx.doi.org/10.1007/s00603-008-0005-1]
- [8] M. Cai, H. Morioka, P.K. Kaiser, Y. Tasaka, H. Kurose, M. Minami, and T. Maejima, "Back-analysis of rock mass strength parameters using AE monitoring data", *Int. J. Rock Mech. Min. Sci.*, vol. 44, pp. 538-549, 2007. [http://dx.doi.org/10.1016/j.ijrmms.2006.09.012]
- [9] A.J. Li, R.S. Merifield, and A.V. Lyamin, "Stability charts for rock slopes based on the Hoek–Brown failure criterion", *Int. J. Rock Mech. Min. Sci.*, vol. 45, pp. 689-700, 2007.
 [http://dx.doi.org/10.1016/j.ijrmms.2007.08.010]
- [10] L.Q. Zhang, Z.Q. Yue, Z.F. Yang, J.X. Qi, and F.C. Liu, "A displacement-based back-analysis method for rock mass modulus and horizontal in situ stress in tunneling-Illustrated with a case study", *Tunn. Undergr. Space Technol.*, vol. 21, pp. 636-649, 2006. [http://dx.doi.org/10.1016/j.tust.2005.12.001]
- [11] S.L. Huang, X.T. Feng, and C.Q. Zhang, "Study of method of comprehensive evaluation for parameters of constitutive model of rock mass", *Chin. J. Rock Mech. Eng.*, vol. 27, pp. 2624-2630, 2008.
- [12] C.X. Liu, L.D. Yang, and W.G. Cao, "A statistical damage softening constitutive model for rock and back analysis of its parameters", *Chin. J. Undergr. Space Eng.*, vol. 3, pp. 453-457, 2007.
- [13] R. Leszek, Computational Intelligence: Methods and Techniques., Springer Press, 2008.
- [14] X.H. Diao, Y.J. Wang, X.T. Fend, and T.W. Chen, ""Identifying mechanics parameters of rock mass by using artificial neural network", J. Northeast. Univ", *Nat. Sci.*, vol. 23, pp. 60-63, 2002.
- S.K. Zhang, and S.D. Yin, "Reservoir geomechanical parameters identification based on ground surface movements", *Acta Geotech.*, vol. 8, pp. 279-292, 2013.
 [http://dx.doi.org/10.1007/s11440-012-0196-1]
- [16] I. Ylmaz, and A.G. Yuksek, "An example of artificial neural network application for indirect estimation of rock parameters", *Mec. Roches*, vol. 41, pp. 781-795, 2008.
- [17] M. Yazdani, M. Sharifzadeh, K. Kamrani, and M. Ghorbani, "Displacement-based numerical back analysis for estimation of rock mass parameters in Siah Bisheh powerhouse cavern using continuum and discontinuum approach", *Tunn. Undergr. Space Technol.*, vol. 28, pp. 41-48, 2011. [http://dx.doi.org/10.1016/j.tust.2011.09.002]
- [18] M. Ghorbani, and M. Sharifzadeh, "Long term stability assessment of Siah Bisheh powerhouse cavern based on displacement back analysis method", *Tunn. Undergr. Space Technol.*, vol. 24, pp. 574-583, 2009. [http://dx.doi.org/10.1016/j.tust.2009.02.007]
- J.H. Wu, and P.H. Tsai, "New dynamic procedure for back-calculating the shear strength parameters of large landslides", *Eng. Geol.*, vol. 123, pp. 129-147, 2011.
 [http://dx.doi.org/10.1016/j.enggeo.2011.01.010]
- [20] C. Rechea, S. Levasseur, and R. Finno, "Inverse analysis techniques for parameter identification in simulation of excavation support systems", *Comput. Geotech.*, vol. 35, pp. 331-345, 2008. [http://dx.doi.org/10.1016/j.compgeo.2007.08.008]
- [21] S. Mostafa, S. Mohsen, and D. Mohammad, "Back analysis of an excavated slope failure in highly fractured rock mass: the case study of Kargar slope failure (Iran)", *Environ. Earth Sci.*, vol. 60, pp. 183-192, 2010. [http://dx.doi.org/10.1007/s12665-009-0178-2]
- [22] Z.H. Zhang, C.B. Zhou, Z.Q. Xiao, G.J. Miao, D.B. Zhang, and N. Jiang, "Sensitivity analysis and orthogonal backward analysis of soil parameters for subway tunnel", J. Cent. South Univ., vol. 44, pp. 2488-2493, 2013.
- [23] D. Stead, E. Eberhardt, and J.S. Coggan, "Developments in the characterization of complex rock slope deformation and failure using numerical modelling techniques", *Eng. Geol.*, vol. 83, pp. 217-235, 2006. [http://dx.doi.org/10.1016/j.enggeo.2005.06.033]
- [24] Y.C. Wang, Y.Q. Shang, and X.H. Xu, "Orthogonal back-analysis of geotechnical parameters and working state evaluation of lining of shallow buried tunnel", J. Cent. South Univ., vol. 42, pp. 1764-1771, 2011.
- [25] Q.S. Zhang, Y.F. Gao, and S.C. Li, "Study on 3D-displacement inversion method for mechanical parameters of mine overlying rock", Met. Mine, vol. 9, pp. 26-31, 2005.
- [26] P.A. Cundall, FLAC User Manual., Itasca Consulting Group Inc: USA, 1993.

- [27] C.B. Zhou, L.W. Guo, Y.K. Yao, X.P. Yin, and S.J. Jiang, "Numerical simulation of wall rock deformation mechanism of mining tunnel", *Rock Soil Mech.*, vol. 30, pp. 654-658, 2009.
- [28] G. Zhang, and W.S. Zhu, "Susceptibility analyses of parameters and optimization of test program", Rock Soil Mech., vol. 14, pp. 51-58, 1993.
- [29] D.E. Rumelhart, G.E. Hinton, and R.J. Williams, Learning Internal Representation by Error Propagation Parallel Distributed Processing: Exploration in the Micro- Structure of Cognition., vol. Vol. 1. MIT Press: Cambridge, MA, 1986.
- [30] F. Jin, Neural computing intelligence: Basic principle, method. Chengdu., Southwest Jiaotong University Press: China, 2000.

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