

A Photogrammetric Approach for Stability Analysis of Weathered Rock Slopes

Dong Hyun Kim, Ivan Gratchev & Arumugam Balasubramaniam

Geotechnical and Geological Engineering

An International Journal

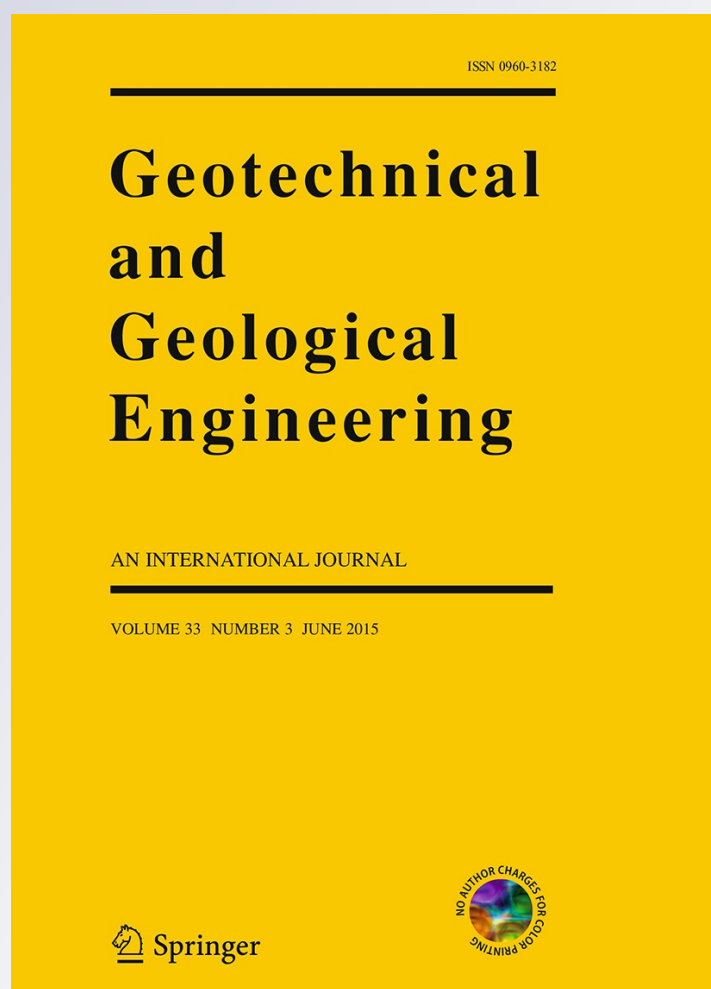
ISSN 0960-3182

Volume 33

Number 3

Geotech Geol Eng (2015) 33:443-454

DOI 10.1007/s10706-014-9830-z



Your article is protected by copyright and all rights are held exclusively by Springer International Publishing Switzerland. This e-offprint is for personal use only and shall not be self-archived in electronic repositories. If you wish to self-archive your article, please use the accepted manuscript version for posting on your own website. You may further deposit the accepted manuscript version in any repository, provided it is only made publicly available 12 months after official publication or later and provided acknowledgement is given to the original source of publication and a link is inserted to the published article on Springer's website. The link must be accompanied by the following text: "The final publication is available at link.springer.com".

A Photogrammetric Approach for Stability Analysis of Weathered Rock Slopes

Dong Hyun Kim · Ivan Gratchev ·
Arumugam Balasubramaniam

Received: 3 October 2013 / Accepted: 15 October 2014 / Published online: 29 October 2014
© Springer International Publishing Switzerland 2014

Abstract The geological strength index (GSI) system is dependent on the rock block volumes and the joint surface conditions. The weathering degree of rock slopes and their strength properties also depend on these characteristics. This study thus focuses on the use of the GSI system and the Hoek–Brown strength criterion to estimate the engineering parameters for weathered rock masses. Photogrammetric methods based on 3D surface models are used to obtain reliable data on the joint sets in rock slopes, instead of general site investigation using labor-intensive techniques. Photogrammetric surveys were conducted on weathered rock slopes in Gold Coast, Australia to obtain the joint spacing, orientation and roughness. The 3D models are then used to estimate the block sizes the joint roughness coefficients (JRC). The block volumes and JRC values were then used to estimate GSI values. Then parametric studies using the finite element method is conducted to investigate the stability of the slope using the GSI values.

Keywords Photogrammetry · Joint spacing · Joint roughness coefficient · Weathering · GSI · Slope stability

1 Introduction

This study seeks to estimate GSI values using photogrammetry to assess the stability of weathered rock slopes in the Tamborine mountain area, Gold Coast, Australia. These slopes have long experienced stability issues affecting the serviceability of the adjacent road. The GSI system is mainly used in the studies that follow a combination of the approaches of Hoek and Brown (1997), Ehlen (1999), Palmström (2001), Cai and Kaiser (2006) and Admassu et al. (2012).

The geological strength index (GSI) system introduced by Hoek and Brown (1997), is a powerful system of rock mass characterization which concentrates on the extent of blockiness and the surface condition of discontinuities. As the GSI values are estimated from visual observation of exposed outcrops, the use of GSI is subjective and requires much experience. Thus, as an effort to quantify GSI values for engineering purposes, Cai and Kaiser (2006) proposed a quantitative method to estimate GSI using relationships between joint roughness number (J_R), joint alteration number (J_A) and block volume (V_b) in three dimensional space.

The general way to obtain the joint properties is to observe rock slopes or boring logs. However, the traditional method may be time consuming and typically associated with risks during field works. To address this problem, the recent development of remote sensing techniques such as photogrammetry, has enabled the accurate estimation of the scale of rock

D. H. Kim (✉) · I. Gratchev · A. Balasubramaniam
Griffith School of Engineering, Griffith University,
Gold Coast, QLD, Australia
e-mail: donghyun.kim@griffith.edu.au

block and orientation of the main joint sets. The technique also provides the 2D profiles and joint roughness coefficient (JRC) from 3D surface models using high resolution 3D digital images, for inaccessible rock slopes (Haneberg 2007; Poropat 2009; Guo et al. 2011; Kim et al. 2013).

During the weathering process, the rock mass becomes decomposed and the number of joint sets may be increased. Thus, the degree of weathering in blocky rock mass can be investigated by measuring the degree of jointing, which is related to the number of joint sets and joint spacing. However, this approach should be carefully considered due to the fact that the frequency of joints can also be caused by different geological processes such as tectonic disturbances (Marinos et al. 2005).

Block size and joint spacing are the main parameters to assess the degree of jointing in rock masses. Palmström (2001) presented correlations between block sizes and joint frequency (J_v) and also RQD and J_v using numerous measurement data. In terms of weathering, Ehlen (1999) showed the correlation between mean joint spacing and weathering grades. Similarly, Admassu et al. (2012) suggested that joint spacing is an important factor contributing to slope failure.

The slake durability and Schmidt rebound values can be used to estimate the strength of the weathered rock joints. Dick et al. (1994) noted that the durability of most rock is one of the most important parameters to the stability of the slopes. The slake durability test is to evaluate the influence of weathering on rocks by measuring the resistance to degradation and decomposition as simulated by being exposed to wetting and drying cycles. The study of Franklin and Chandra (1972) reported that slaking of rocks is an important consideration to evaluate the engineering behavior of rock mass and rock materials in geotechnical practices. Also, Schmidt hammer rebound values showed reliable results to classify weathering in the less weathered material (Arikan et al. 2007). Due to the reduction of strength of the weathered rock surface, rebound energy of Schmidt hammer can be absorbed. Sharma et al. (2011) showed a linear relation between the Schmidt rebound number and the impact strength index (ISI) and the slake durability index (SDI).

To investigate the behavior of rock masses for different weathering conditions, finite element analyses were also performed using the Hoek–Brown criterion based on the calculated GSI values.

2 Estimation of the Degree of Jointing

In principle, weathering causes lower values in the spacing of the joints and intact rock strength. As rock surfaces are exposed for a long time, the rock breaks into small pieces and the number of joint sets becomes visible. Hack and Price (1997) investigated the influence of weathering for the parameters of intact rock strength and spacing of discontinuities. Similarly, Ehlen (1999) presented that joint spacing tends to become closer when weathering grades increase. The spacing of discontinuities will determine the dimensions of the rock blocks in the slope and the block volume is an extremely important parameter that is directly related to the rock mass classification systems.

The development of methods to estimate the block volumes has been described extensively in the literature. The correlation between the block volume (V_b) and the volumetric joint count (J_v) is proposed by Palmström (2001),

$$V_b = \beta \times J_v^{-3} \quad (1)$$

where β is the block shape factor.

The block shape factor can be estimated by Eq. (2)

$$\beta \approx 20 + 7a_3/a_1 \quad (2)$$

where a_1 and a_3 are the shortest and longest dimensions of the block.

The volumetric joint count (J_v) is a measure for the degree of jointing or the inter block size and it can be used as an input in the description of rock mass jointing (Palmström 2001). The J_v uses the term, joint spacing (S), which is the distance between an adjacent pair of discontinuities measured along a line on the rock surface. The J_v count is defined as the number of joints intersecting a volume of 1 m³,

$$J_v = 1/S_1 + 1/S_2 + 1/S_3 + \dots + 1/S_n, \quad (3)$$

where joint spacing (S) is S_1 , S_2 and S_n are the average spacing for the joint sets.

In three dimensional space, individual block volumes can also be estimated using joint spacing and their orientations. Assuming that the main joint sets were persistent, the block volume is calculated using Eq. (4) proposed by Cai et al. (2004),

$$V_b = \frac{s_1 s_2 s_3}{\sin \gamma_1 \sin \gamma_2 \sin \gamma_3} \quad (4)$$

where s_1, s_2, s_3 = spacing between discontinuity sets;
 $\gamma_1, \gamma_2, \gamma_3$ = angles between discontinuity sets.

3 Quantification of GSI Values

The GSI is the geological index based on rock mass structures and joint surface conditions. Marininos et al. (2005) suggested that the GSI values in weathering condition can be estimated by shifting the values of the unweathered rock mass in the chart. However, this procedure tends to be subjective and requires long term experience. As the GSI values are estimated from visual observation of exposed outcrops, the GSI values for weathered rock masses can be assessed using the degree of jointing on the slope surfaces. Cai and Kaiser (2006) proposed a quantitative method to estimate GSI using the joint condition factor (J_c) and rock block volume (V_b) in three dimensional space. The GSI is defined from the following function,

$$GSI = \frac{26.5 + 8.79 \ln J_c + 0.9 \ln V_b}{1 + 0.0151 \ln J_c - 0.0253 \ln V_b} \quad (5)$$

where V_b is the block volume and J_c is a quantitative characterization factor to show joint condition. The joint condition factor, J_c is defined as follows;

$$J_c = \frac{J_W J_S}{J_A} \quad (6)$$

where J_W is a factor to describe large scale waviness of joints which is in meters from 1 to 10 m and J_S is a term to describe small scale smoothness. J_s is a parameter to describe joint surface roughness which is closely related to the joint roughness coefficient (JRC) defined by Barton et al. (as cited in Palmström 2001). J_A is the joint alteration factor which defines the filling of the joints.

As a main parameter, GSI values are applied in the Hoek–Brown failure criterion. The normal form of the Hoek–Brown failure criterion is Eq. (7),

$$\sigma'_1 = \sigma'_3 + \sigma_{ci} \left(m_b \frac{\sigma'_3}{\sigma_{ci}} + s \right)^\alpha \quad (7)$$

where σ'_1 and σ'_3 are the major and minor effective principal stresses and σ_{ci} is the uniaxial compressive strength of the rock. m_b, s and α are material constants for rock mass as follows;

$$m_b = m_i \exp \left(\frac{GSI - 100}{28 - 14D} \right) \quad (8)$$

$$s = \exp \left(\frac{GSI - 100}{9 - 3D} \right) \quad (9)$$

$$\alpha = \frac{1}{2} + \frac{1}{6} \left(e^{-\frac{GSI}{15}} - e^{-\frac{20}{3}} \right) \quad (10)$$

The deformation modulus in the GSI system can be estimated as follows;

$$E = \left(1 - \frac{D}{2} \right) \sqrt{\frac{\sigma_c}{100}} 10^{\left(\frac{GSI - 10}{40} \right)} \text{ GPa} \quad (11)$$

(for $\sigma_c < 100$ MPa)

where D is the disturbance factor that depends on the degree of disturbance by blasting and stress relaxation (Hoek et al. 2002). For weathered rock masses, the constants m_i and the unconfined strength of intact rock σ_{ci} in the Hoek–Brown criterion should be reduced in comparison with the unweathered rock masses (Marininos et al. 2005).

4 Geological Conditions of the Study Area

4.1 Geology

The study areas are located on the eastern road near Tamborine mountain that is mainly composed of layered argillite and sandstone in Neranleigh–Fernvale Beds (Willmott 2010; Gratchev et al. 2013; Shokouhi et al. 2013) (Fig. 1). In the study area, the rocks are heavily weathered, folded and steeply inclined. Argillite, which is hardened and slightly recrystallized shale, is fined-grained rock, bedding, and fractured in many exposures. The sandstone is mostly coarse-grained sediment of bright brown color. Recently, the slope experienced failures during heavy rainfall. It was observed from the site investigation that the rock beneath the failed area was deeply weathered and the bedding structure associated with other discontinuities was significant factor contributing to the failure.

4.2 Strength Properties of Rock Mass

A series of point load tests and slake durability tests were performed on the both collected samples and

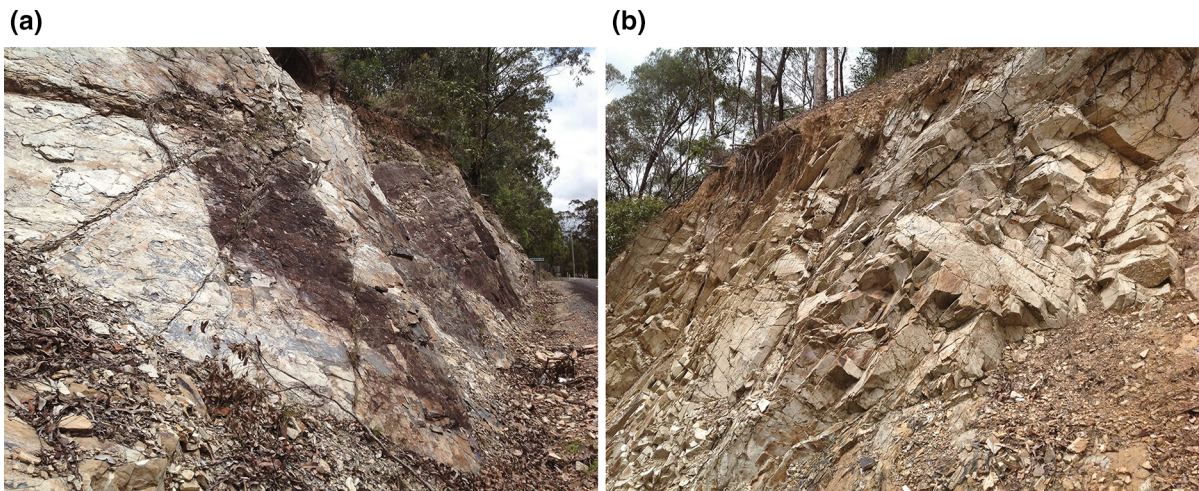
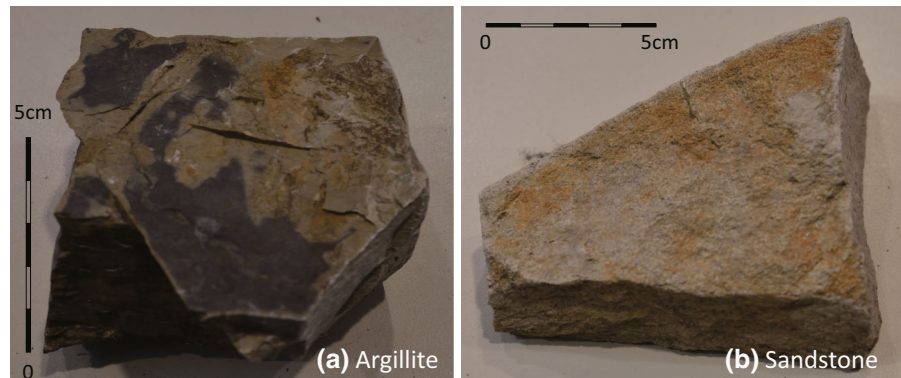


Fig. 1 Geological conditions of study area, argillite (a) and sandstone (b)

Fig. 2 Collected rock samples



core samples from the site (Fig. 2). The core samples from near sites were used as unweathered rocks which were not exposed to surface weathering. Figure 3 shows the range of point load index (PLI) and the slake durability indices (SDI) of the rock samples. The samples were classified by the point load strength (Broch and Franklin 1972). A large number of weathered argillite samples exist in the ‘high’ strength category of PLI while the weathered sandstone samples are categorized as the ‘medium’ PLI.

The unconfined compressive strengths (UCS) were estimated from the results of the point load tests. The approximate conversion of UCS was performed using the conversion factor 24 as shown in Eq. (12) (Broch and Franklin 1972),

$$UCS = 24 \times I_s(50) \quad (12)$$

where, $I_s(50)$ is the point load strength index at a reference diameter of 50 mm for core samples. The laboratory tests also show that the samples with low values of unconfined compressive strength tend to have low values of slake durability index. The slake durability indices from the argillite samples range from 92.5 to 97 % and the point load strength indices range from 1.53 to 3.31 MPa. The durability of argillite is classified as ‘high’ to ‘extremely high’ durability according to the slake durability index classification (Franklin and Chandra 1972). In contrast, a considerable difference in slake durability index (Id_2) of the sandstone was indicated between ‘low to medium’ ($Id_2 = 50$ %) and ‘very high’ durability ($Id_2 = 90.7$ %). As a result, the argillite of the study area has higher unconfined compressive strength and more durability than the sandstone (Table 1).

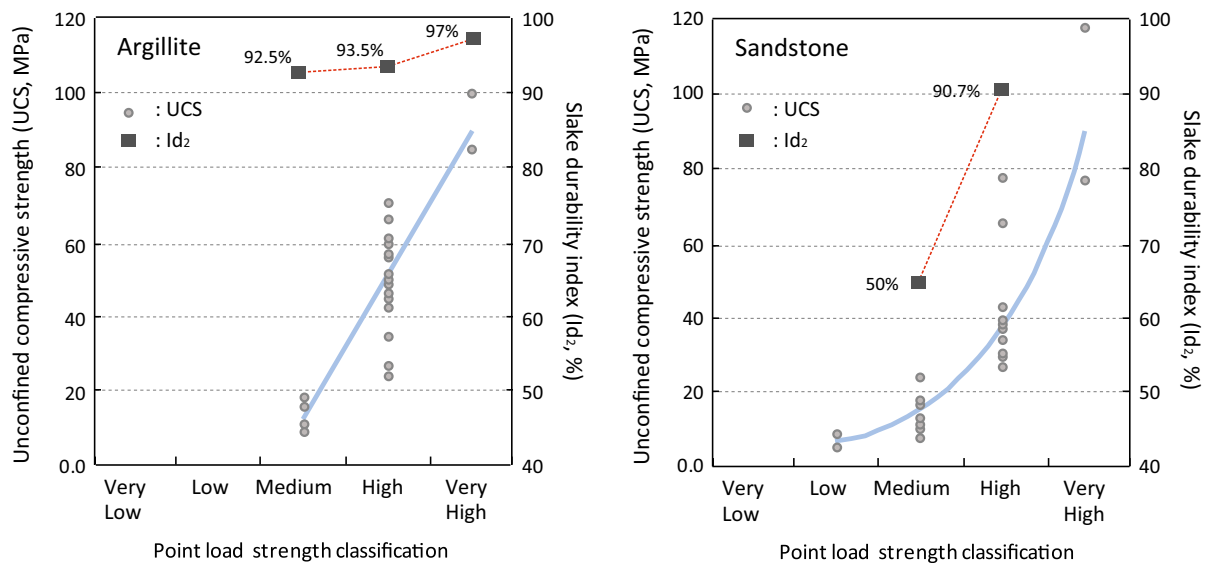


Fig. 3 Unconfined compressive strength (UCS) and slake durability index (Id_2) according to point strength classification

Table 1 Unconfined compressive strength based on point load tests

Rock types	Fresh				Weathered			
	Test number	UCS (MPa)			Test number	UCS (MPa)		
		Min	Max	Mean		Min	Max	Mean
Argillite	5	85.3	100.1	92.7	23	9.4	70.6	41.0
Sandstone	15	22.9	77.6	46.4	12	6.0	16.2	12.7

Table 2 Results of Schmidt hammer tests

Rock types	Schmidt hammer rebound			Weathering grade
	Range of data	Mean	SD	
Argillite	17–49	31.3	8.3	III (20–30)
Sandstone	11–46	27.7	10.9	IV (30–40)

During the site investigations, a total of 40 Schmidt hammer tests were performed on the discontinuities of the argillite and sandstone areas. Table 2 shows that the range of rebound values are between 11 and 49, which are classified as weathering grade III or IV for the both argillite and the sandstone areas (Arikan et al. 2007). The different weathering grades also indicate that the sandstone of the study area is less durable than the argillite.

5 Assessment of Joint Properties Using Photogrammetry Methods

5.1 Photogrammetry Survey

Photogrammetry survey was performed to produce 3D models of the slope surface model and determine the characteristics of discontinuities. A professional digital camera (Nikon D7000) and lens of 24 mm focal length lens was employed to capture high-quality digital images of the slope at two camera positions. The computer code “Sirovision” (CSIRO 2005) was used to analyse the images and create 3D models of each slope site. Georeferencing was performed for the photo by determining the coordinates of the left camera position (using a GPS device), and measuring its bearing (azimuth) to the centre of slope (using a geological compass) (Sturzenegger 2010).

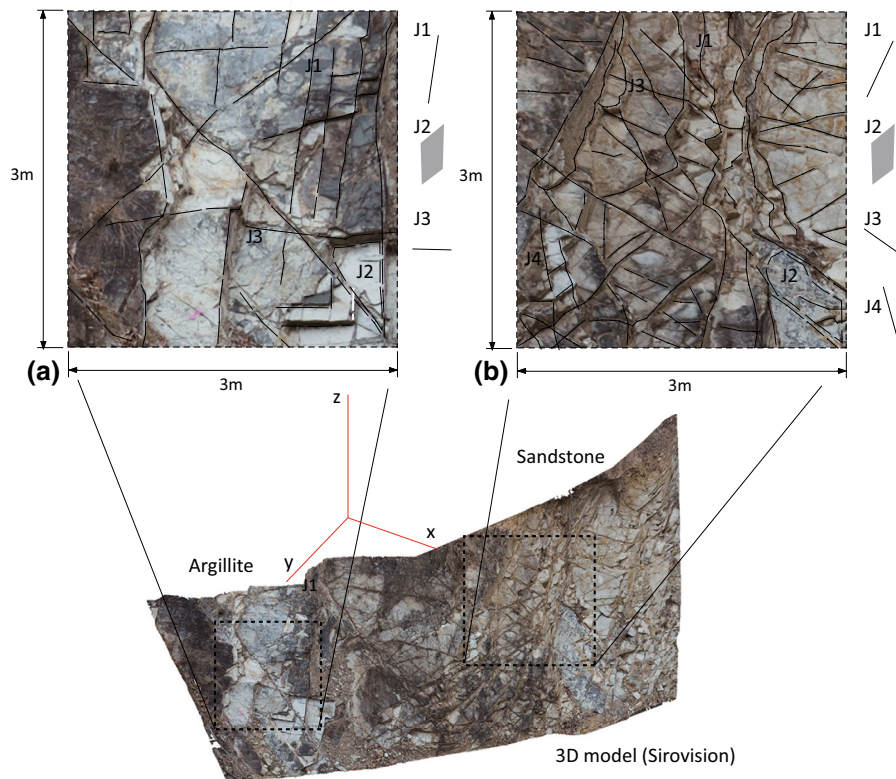


Fig. 4 Main joint sets of the selected areas based on the 3D model (Sirovision)

5.2 Degree of Jointing

The spacing and the orientations of joints were obtained from the 3D models. Mapping and tracing for the joint information was conducted manually on screen and provided the spacing and the length of the joint sets. The discontinuities data obtained from the 3D model were plotted on stereonets to determine the main joint sets. For the both argillite and sandstone, representative sampling areas, which are 3 m × 3 m in dimension, were considered in this study (Fig. 4).

The block shapes in the sandstone area are more irregular than those of argillite (Fig. 4). Figure 5 shows the joint spacings of the main joint sets for the argillite and the sandstone. The overall joint spacing of the sandstone area is smaller than that of the argillite area. The measured distances and orientations of the discontinuity sets were used to assess the degree of jointing; block volumes, volumetric joint count and RQD. As a result, the block volumes and RQD values of argillite indicates higher values than those of

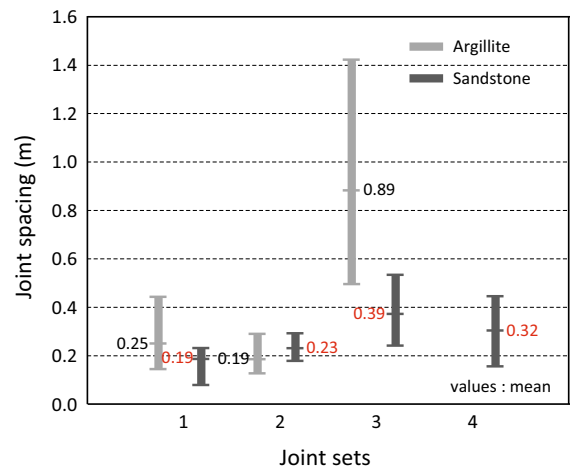


Fig. 5 Variations in the spacing of joints

sandstone as presented in Table 3. Even though the block sizes of both rock types can be formed based on different geologic processes such as tectonic movement, it is interesting that the sandstone area shows

Table 3 Estimated degree of jointing in the study areas based on 3D models

Degree of jointing	Argillite	Sandstone
Volumetric joint count (J_v)	6.4–16.4 Mean 10.5	11.6–22.1 Mean 15.3
Block volume (V_b , $\times 10^3 \text{ cm}^3$)	7.0–113.0 Mean 26.0	3.0–19.0 Mean 8.0
RQD (%)	61.0–93.8 Mean 80.4	42.1–76.8 Mean 64.7

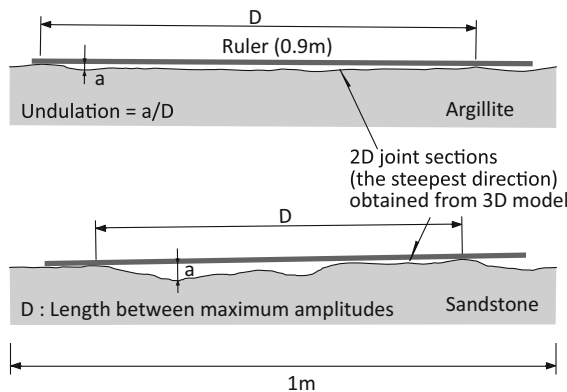


Fig. 6 Estimation of undulation using 3D models

lower block volumes with lower weathering degrees (IV) than that of argillite (III) (See Table 2).

5.3 Estimation of GSI Values

The waviness of joint wall (J_w) was estimated using undulations obtained by the roughness profiles extracted from the 3D models. The procedure can be applied with a 0.9 m long edge which is placed on the

profiles as shown in Fig. 6. The concept of the measurement is the same as the procedure described by Piteau (1970).

In a small scale roughness, the joint smoothness factor (J_s) can be associated with JRC values. In this study, J_s values were estimated using the linear interpolation relationship between JRC values and J_s [Eq. (13)]. This relationship is based on the assumption that the range of J_s (0.6–3.0) is directly proportional to the range of JRC (0–20).

$$J_s = 0.1125 \times JRC + 0.75 \quad (13)$$

The JRC values were obtained using the Maerz et al. (1990) and Tse and Cruden (1979) approaches based on the surface profiles in 10 cm lengths which were extracted from the 3D models for the dip directions of the joints. 20 JRC values were calculated at different locations in each section. Figure 7 presents an example of roughness profiles extracted from the 3D model. JRC values were calculated using the code Sirovision with functions as follows,

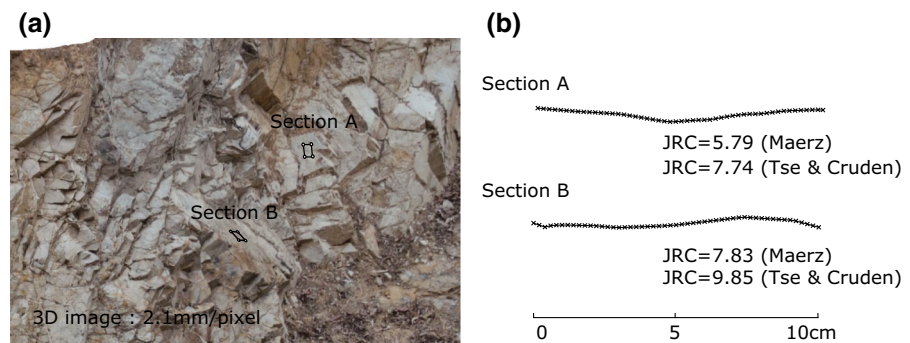
$$JRC = 32.2 + 32.47 \log Z_2 \quad (\text{Tse and Cruden 1979}) \quad (14)$$

$$JRC = 411(R_p - 1) \quad (\text{Maerz et al. 1990}) \quad (15)$$

where Z_2 is a roughness parameter using variances within asperity heights, and where R_p is a roughness parameter which is related to the inclination angle (i) of the sawtooth surface of the roughness profile.

Figure 8 summarizes the obtained data, indicating a similar range of JRC values between both the argillite and sandstone areas from 7.4 to 10.3. Consequently, there is no significant difference between the mean values of J_s ; however, in the large scale joint condition, the more undulating surfaces of the

Fig. 7 JRC values estimated from 3D models, 3D image (a) JRC values and roughness profiles (b)



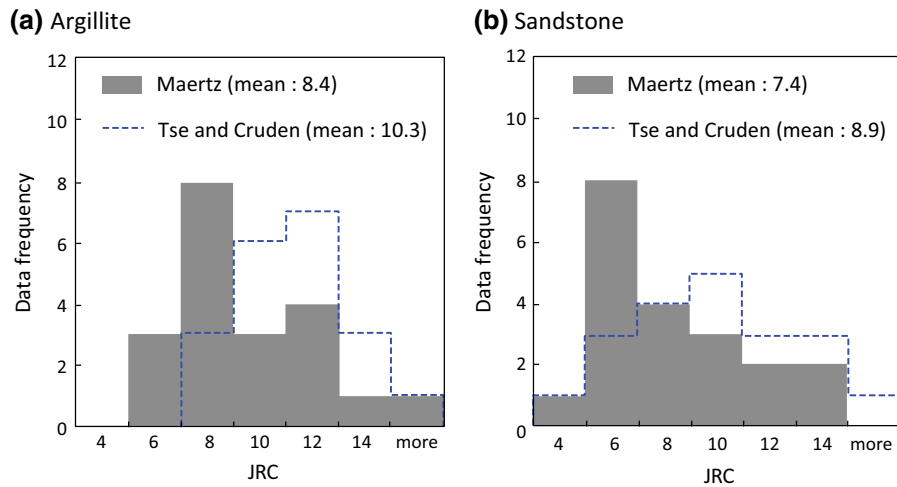


Fig. 8 Distribution of JRC data obtained from 3D image

Table 4 Estimated joint surface properties based on 3D models

Joint surface conditions	Argillite	Sandstone
Waviness of joints (J_w)	1.5	2.0
Smoothness factor (J_s)	1.7–1.9	1.6–1.8
	Mean 1.8	Mean 1.7
Joint condition factor (J_c)	0.64–0.72	0.79–0.88
	Mean 0.68	Mean 0.83

sandstone produced larger J_c values than the argillite section, as presented in Table 4.

The GSI values were obtained from Eq. (5) (Cai and Kaiser 2006). The values were quantified by the estimated block volumes and joint condition factors

(J_c). The variation of GSI as a function of block volume (V_b) and joint condition factor (J_c) are plotted in Fig. 9. As shown in Fig. 9, the block volume is a less sensitive parameter to estimate GSI values than the joint condition factor. Figure 9a shows that even much larger blocks (100 times) with the same joint condition factors produce an increase only 10 GSI values. Therefore, joint condition factors can be more influential and thus should be carefully examined to quantify GSI values. For the study areas, the estimated GSI values and the Hoek–Brown strength parameters were given in Table 5. The numerical code ‘RocData’ (Rocscience) was used to estimate the Hoek–Brown strength parameters and the elastic modulus.

Fig. 9 GSI variation with V_b (a) and J_c (b) based on Eq. (5)

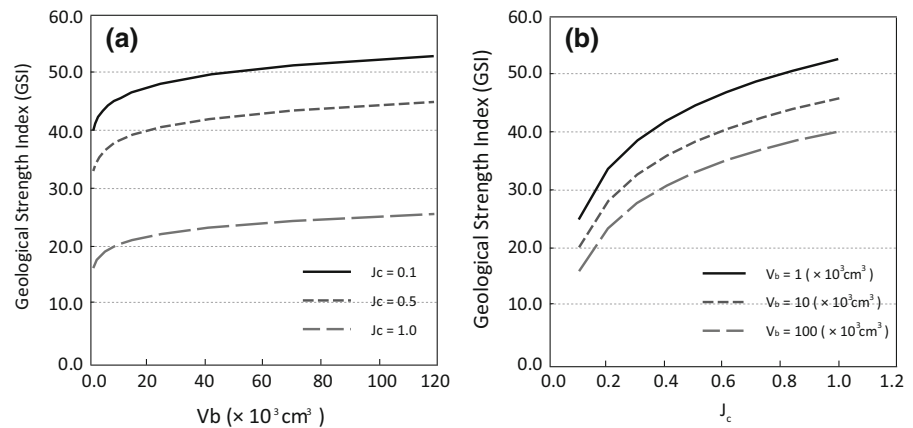


Table 5 Estimated GSI values and Hoek–Brown properties using RocData (Rocscience)

Strength properties	Argillite	Sandstone
UCS (MPa)	41.0	12.7
Hoek–Brown properties		
GSI	44	43
M_b	0.83	0.742
s	0.0003	0.0003
a	0.509	0.509
Young's modulus (GPa)	1.11	0.6

6 Parametric Study on the Weathered Rock Slope

Based on the estimated GSI values and the block sizes obtained from the 3D models, the parametric analyses were performed using a numerical code, Phase2 (Rocscience). This code is an elasto-plastic finite element program for calculating stresses and displacements of rock slopes, and also provides joint simulations. The failure mechanisms of the study area were simulated by two different criteria in numerical analyses. One was to investigate the influence of the block sizes on the behaviour of weathered slopes using the Mohr–Coulomb model, and the other was to examine the influence of GSI values on the failure behaviour for the same analysis section. Figure 10 demonstrates the numerical models for both cases.

Firstly, the numerical models using the Mohr–Coulomb criterion have one main joint set dipping 43° along the slope and another cross-joint set generated perpendicular to the slope. In the parametric study, three different joint spaces—10, 20 and 40 for the main joint set and 20, 40 and 80 for the cross-joint set—were considered in order to simulate different

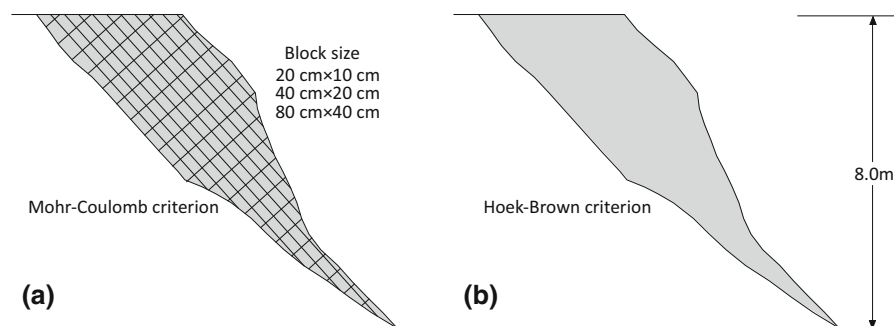
Table 6 Conditions of numerical analysis (strength criteria and estimated parameters)

Models	Criteria and parameters	Argillite	Sandstone
1	Mohr–Coulomb criterion (rock mass)	Cohesion: 0.13 MPa	Cohesion: 0.07 MPa
		Friction angle: 55.0°	Friction angle: 47.0°
	Barton–Bandis criterion (joints)	JRC: 9.4	JRC: 8.2
		JCS: 41 MPa	JCS: 12.7 MPa
2	Block size	0.002, 0.008, 0.032 (m^3)	0.002, 0.008, 0.032 (m^3)
	Hoek–Brown criterion		
	GSI	20, 40, 60	20, 40, 60
	Material constants, m_b	0.22–1.99	0.21–1.88
	Material constants, s	9.2e–6–0.003	9.2e–6–0.003
	Material constants, a	0.54–0.503	0.54–0.503

block sizes (Fig. 10a). For the joint parameters, the Barton–Bandis rock joint strength criterion was adopted and the peak friction angle which represents joint roughness condition, was estimated using the JRC values of Eq. (16) suggested by Barton and Choubey (1977),

$$\phi = \phi_r + JRC_n \log_{10} \left(\frac{JCS_n}{\sigma_n} \right) \quad (16)$$

where, ϕ_r = residual friction angle of joint; JCS = joint wall compressive strength; σ_n = normal stress acting on the joint plane.

**Fig. 10** Numerical models for the parametric study using Phase 2, Mohr–Coulomb model (a), Hoek–Brown model (b)

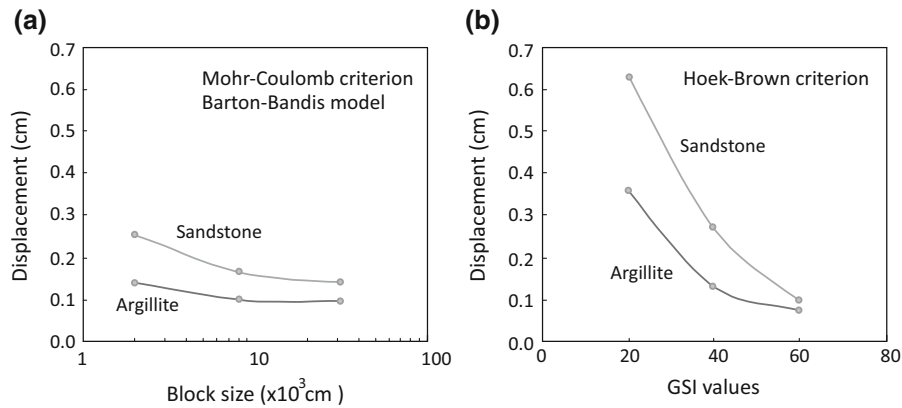


Fig. 11 Variation of maximum displacement with block volumes (a) and with GSI values (b)

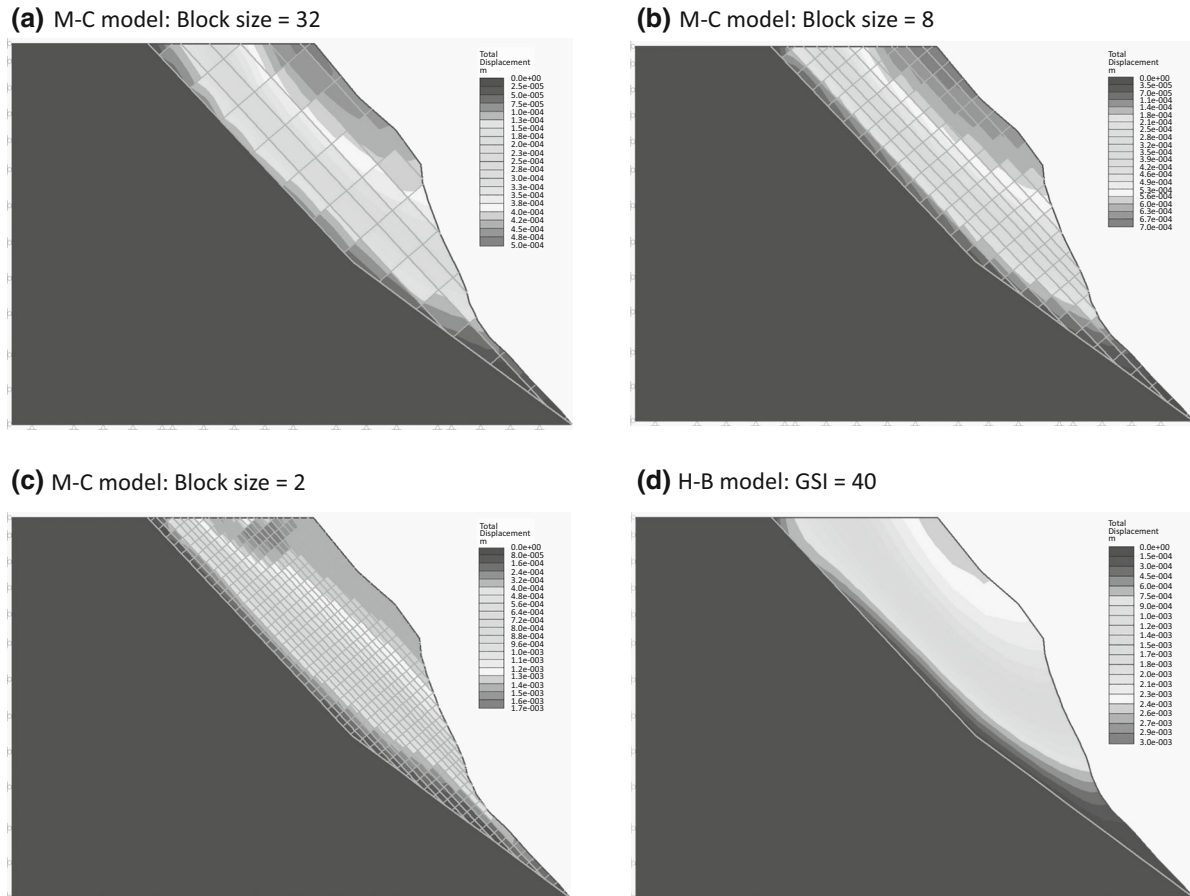


Fig. 12 Contours of total displacement (sandstone), Mohr-Coulomb model (a–c), Hoek-Brown model (d)

Secondly, the numerical models of the Hoek-Brown failure criterion were simulated for the same failure area. Based on three different GSI values,

corresponding material constants described in Eqs. (8, 9, 10) were estimated by 'RocData'. The conditions for both numerical analyses are summarized in Table 6.

Using the results of FEM analysis, slope failure mechanisms can be inferred from the assessment of total displacement. Figure 11 shows the total displacements from the parametric analyses of the slope models. Figure 11a indicates that the variation in block size can affect the deformation of the slopes. As the block sizes increase, the deformation values diminish. However, increasing the block size does not significantly reduce the displacement of the slope. This is mainly because there are no other changes in the strength parameters except for block sizes. Based on the JRC values obtained from the 3D image, block size itself is not an influential factor for the study slope.

On the other hand, the results analysed from the Hoek–Brown criterion shows more sensitive variations of slope displacement. Figure 11b shows total displacement of the slope according to the variations of GSI values and material constants. The results of the analyses demonstrated that when the GSI value increased from 20 to 60, the maximum value of total displacement of the slope was considerably reduced. This means that the GSI values which are controlled by the block sizes and by joint conditions can affect strength characteristics in the Hoek–Brown criterion. Consequently, in numerical analyses, the block sizes are embedded in the GSI values, resulting in an effect on the behaviour of slopes with corresponding strength properties. Figure 12 shows the deformed mesh and the total displacement contours for each analysis.

7 Conclusion

The photogrammetry method was applied to investigate the stability of a weathered rock slope. Based on the obtained results, the following conclusions can be drawn:

- Photogrammetry can provide reliable data on joint spacing and the block sizes of rock slopes based on the discontinuity sets in 3D models.
- In the study area, sandstone showed a higher weathering grade (IV) than argillite (III) and the block size of the slope can be an indicator of the weathering degree for the rock slopes.
- The 3D models can produce joint roughness coefficients (JRC) and these values were effectively used to estimate the joint smoothness factors (J_s) that were essential in obtaining GSI values.

- In the quantification of GSI values using the function suggested by Cai and Kaiser (2006), joint condition factor (J_c) can be a more sensitive parameter than block sizes. In addition, block size itself was not a sensitive parameter in the results of the FEM analysis.
- Rock block sizes can be effectively considered using GSI systems, combined with photogrammetry 3D models, to estimate the corresponding strength parameters for the Hoek–Brown criterion for weathered rock slopes.

Acknowledgments The authors wish to thank Mr. George Poropat from CSIRO for the valuable assistance.

References

- Admassu Y, Shakoor A, Wells NA (2012) Evaluating selected factors affecting the depth of undercutting in rocks subject to differential weathering. *Eng Geol* 124:1–11
- Arikan F, Ulusay R, Aydin N (2007) Characterization of weathered acidic volcanic rocks and a weathering classification based on a rating system. *Bull Eng Geol Environ* 66:415–430
- Barton N, Choubey V (1977) The shear strength of rock joints in theory and practice. *Rock Mech* 10:1–54
- Broch E, Franklin JA (1972) The point load strength test. *Int J Rock Mech Min Sci* 9:669–697
- Cai M, Kaiser PK (2006) Visualization of rock mass classification systems. *Geotech Geol Eng* 24:1089–1102
- Cai M, Kaiser PK, Uno H, Tasaka Y, Minami M (2004) Estimation of rock mass deformation modulus and strength of jointed hard rock masses using the GSI system. *Int J Rock Mech Min Sci* 41:3–19
- CSIRO (2005) Field procedures for photogrammetric pit mapping. CSIRO Exploration and Mining
- Dick JC, Shakoor A, Wells N (1994) A geological approach toward developing a mudrock-durability classification system. *Can Geotech J* 31:17–27
- Ehlen J (1999) Fracture characteristics in weathered granites. *Geomorphology* 31:29–45
- Franklin JA, Chandra A (1972) The slake durability test. *Int J Rock Mech Min Sci* 9:325–341
- Gratchev I, Shokouhi A, Kim D, Stead D, Wolter A (2013) Assessment of rock slope stability using remote sensing technique in the Gold Coast area, Australia. In: *Proceedings of the 18th Southeast Asian geotechnical and inaugural AGSSEA conference*, pp 729–734
- Guo H, Karekal S, Poropat G, Soole P, Lambert C (2011) Pit wall strength estimation with 3D imaging. CSIRO, ACARP, Brisbane
- Hack R, Price D (1997) Quantification of weathering. In: *Proceedings of engineering geology and the environment*, Athens, Balkema, Rotterdam, pp 145–150

- Haneberg WC (2007) Directional roughness profiles from three-dimensional photogrammetric or laser scanner point clouds. In: Proceedings of the 1st Canada–US rock mechanics symposium, Vancouver, pp 101–106
- Hoek E, Brown ET (1997) Practical estimates of rock mass strength. *Int J Rock Mech Min Sci* 34(8):1165–1186
- Hoek E, Carranza-Torres C, Corkum B (2002) Hoek–Brown failure criterion-2002 edition. In: Proceedings of NARMS-TAC conference, Toronto, pp 267–273
- Kim DH, Gratchev I, Balasubramaniam AS (2013) Determination of joint roughness coefficient (JRC) for slope stability analysis: a case study from the Gold Coast area, Australia. *Landslides*. doi:[10.1007/s10346-013-0410-8](https://doi.org/10.1007/s10346-013-0410-8)
- Maerz NH, Franklin JA, Bennett CP (1990) Joint roughness measurement using shadow profilometry. *Int J Rock Mech Min Sci* 27:329–343
- Marinos V, Marinos P, Hoek E (2005) The geological strength index: applications and limitations. *Bull Eng Geol Environ* 64:55–65
- Palmström A (2001) In-situ characterization of rocks. A. A. Balkema Publishers, Lisse/Abingdon/Exton(PA)/Tokio
- Piteau DR (1970) Geological factors significant to the stability of slopes cut in rock. In: Proceedings of the on planning open pit mines, Johannesburg, South Africa, pp 33–53
- Poropat GV (2009) Measurement of surface roughness of rock discontinuities. In: Proceedings of the 3rd CANUS rock mechanics symposium, Toronto, Canada, paper 3976
- Sharma PK, Khandelwal M, Singh TN (2011) A correlation between Schmidt hammer rebound numbers with impact strength index, slake durability index and P-wave velocity. *Int J Earth Sci (Geologische Rundschau)* 100:189–195
- Shokouhi A, Gratchev I, Kim D (2013) Rock slope stability problems in Gold Coast area, Australia. *Int J Geomate* 4(1):501–504
- Sturzenegger M (2010) Multi-scale characterization of rock mass discontinuities and rock slope geometry using terrestrial remote sensing techniques. Ph.D. thesis, Simon Fraser University
- Tse R, Cruden DM (1979) Estimating joint roughness coefficients. *Int J Rock Mech Min Sci* 16:303–307
- Willmott WF (2010) Rocks and landscape of the Gold Coast Hinterland. Geological Society of Australia, Queensland Division