# Estimating the three-dimensional joint roughness coefficient value of rock fractures 

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#### Abstract

Measurement and estimation of the joint roughness coefficient (JRC) is a critical but also difficult challenge in the field of rock mechanics. Parameters for estimating JRC based on a profile derived from a fracture surface are generally twodimensional (2D), where a single or multiple straight profiles derived from a surface cannot reflect the roughness of the entire surface. It is therefore necessary to derive the threedimensional (3D) roughness parameters from the entire surface. In this article, a detailed review is made on 3D roughness parameters along with classification and discussion of their usability and limitations. Methods using Triangulated Irregular Network (TIN) and 3D wireframe to derive 3D roughness parameters are described. Thirty-eight sets of fresh rock blocks with fractures in the middle were prepared and tested in direct shear. Based on these, empirical equations for JRC estimation using 3D roughness parameters have been derived. Nine parameters $\left(\theta_{s}, \theta_{g}, \theta_{2 s}, S_{s T}, S_{s F}, V_{a n}, Z_{s a}, Z_{\mathrm{rms}}\right.$, and $Z_{\text {range }}$ ) are found to have close correlations with JRC and are capable of estimating JRC of rock fracture surfaces. Other parameters ( $Z_{s s}, Z_{s k}, V_{s v i}, V_{s c i}, S_{d r}$ and $\left.S_{t s}\right)$ show no good correlations with JRC. The sampling interval has little influence when using volume and amplitude parameters ( $V_{\mathrm{an}}, Z_{s a}, Z_{r m s}$, and $Z_{\text {range }}$ ) for JRC estimation, while it influences to some extent when other parameters $\left(\theta_{s}, \theta_{g}, \theta_{2 s}, S_{s T}\right.$ and $\left.S_{s F}\right)$ are used. For their easy calculation, the equations with amplitude parameters are recommended to facilitate rapid estimation of JRC in engineering practice.


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Keywords JRC estimation • Empirical equation • 3D roughness parameter • Sampling interval

## Introduction

The rock joint roughness coefficient (JRC) was proposed by Barton (1973) to estimate the peak shear strength of joints using the following empirical equation, which is also called the JRC-JCS model:
$\tau=\operatorname{\sigma tan}\left[J R C l o g(J C S / \sigma)+\varphi_{b}\right]$
where $\tau$ is the peak shear strength of the rock joint, $\sigma$ is the normal stress, JRC is the joint roughness coefficient, JCS is the strength of joint wall, and $\varphi_{b}$ is the basic friction angle.

Measurement and estimation of the JRC is critical for using the JRC-JCS model but also a difficult challenge in the field of rock mechanics (Barton and Bandis 1990). The JRC of a particular rock joint profile is most often estimated by visibly comparing it to the ten standard profiles with JRC values ranging from 0 to 20 (Barton and Choubey 1977). This approach has also been adopted by the ISRM (International Society for Rock Mechanics) Commission on Testing Methods since 1981 (Brown, 1981). However, the visible comparison is subjective since the user has to judge which profile the joint in question fits the best.

The development of objective methods was gradually advanced by researchers considering statistical parameters and the fractal dimension of the rock joint profiles. The most often used parameters include $Z_{2}$ (the root mean square of the first deviation of the profile), $\sigma_{I}$ (standard deviation of the angle I), $R_{z}$ (the maximum height), $\lambda$ (the ultimate slope), $\delta$ (profile
elongation index), $\lambda_{Z 2}$ (directional roughness index), $\beta_{100 \%}$ (average slope angle against shear direction), $D_{c}$ (fractal dimension determined by compass-walking method), and $D_{h-L}$ (fractal dimension determined via hypotenuse leg method). Among these, amplitude parameters ( $R_{z}, \lambda$ and $D_{h-L}$ ) show a lower sensitivity to the sampling interval (SI) than slope ( $Z_{2}$, $\beta_{100 \%}$, and $\sigma_{I}$ ) and elongation parameters ( $\delta$ ) in the determination of two-dimensional (2D) JRC (Li et al. 2016; Zheng and Qi 2016; Liu et al. 2017). Correlations between these parameters and JRC can be found in Tse and Cruden (1979), Yu and Vayssade (1991), Wakabayashi and Fukushige (1992), Tatone and Grasselli (2010), and Zhang et al. (2014), and in the reviews by Li and Zhang (2015), Li and Huang (2015), and Zheng and Qi (2016). However, these correlations are all based on 2D roughness profiles, i.e., cross-sections along straight lines over the joint surface. There are no welldeveloped methods to achieve roughness parameters for the entire fracture surface and no reliable equations for estimating JRC with such parameters.

This study gives a detailed review on parameters describing the roughness of an entire surface along with a classification and discussion about their usability and limitations. Methods using Triangulated Irregular Network (TIN) and three-dimensional (3D) wireframe to derive 3D roughness parameters are proposed. Based on direct shear tests of 38 sets of rock joints, a set of empirical equations are proposed for JRC estimation using 3 D roughness parameters.

## Literature review

A detailed literature review of 3D roughness parameters representing an entire fracture surface is summarized in this section. Most morphological reconstructions of fracture surfaces are realized by the 3D wireframe (curved rectangle) model (Belem et al. 2000; Zhang et al. 2009) or TIN model (Belem et al. 2000; Grasselli 2001; Cottrell 2009; Grasselli and Egger 2003; Lee et al. 2011; Tang et al. 2012). Elements of the 3D wireframe or TIN have their own physical properties including dip angle, dip direction, height, area, etc. Parameters with the same physical significance proposed by different researchers can be classified into four groups of slope, area, volume, and amplitude.

Slope parameters are related to dip angle or apparent dip angle of the elements of 3D wireframe or TIN models. Grasselli (2001) rebuilt the fracture surface by TIN and took apparent dip angle $\left(\theta^{*}\right)$ of the elements against shearing direction and potential area ratio $\left(A_{\theta^{*}}\right)$ of TIN elements to describe the roughness of fracture surface. The area ratio is given by $A_{\theta^{*}}=A_{p i} / A_{t}$, where $A_{p i}$ is the total area of triangular elements against the shear direction with apparent slopes larger than $\theta^{*}$ and $A_{t}$ is the actual area of the
Table 1 Three-dimensional slope parameters describing fracture surface roughness

| Term | Definition | Calculation | Original | Anisotropy | $E^{\text {JRC }}$ | References |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| $\theta_{c i}$ | Slope index of element surfaces facing shearing direction | $\theta_{c}=\theta_{\text {max }}{ }^{\prime}(1+C)$ | $\theta^{*} \max / C ; \theta^{*}{ }_{\text {max }} /(1+C)$ | $\checkmark$ | $\checkmark$ | Grasselli (2001); Cottrell (2009); Grasselli and Egger (2003); Tatone and Grasseli (2010) |
| $\theta_{s}$ | Surface angularity of the entire surface | $\theta_{s}=\frac{1}{n} \sum_{I=1}^{n} \alpha_{i j}$ | $\theta_{s}$ |  | $\checkmark$ | Belem et al. (2000); Lee et al. (2011); Tang et al. 2012) |
| $\theta_{s i}$ | Surface angularity of element surfaces facing shearing direction | $\theta_{s i}=\frac{1}{m_{s}} \sum_{I=1}^{m_{s}} \alpha_{i j, s}$ | $\theta_{s}{ }^{\prime}$ | $\checkmark$ | $\checkmark$ | Lee et al. (2011) |
| $\theta_{2 s}$ | Root mean square of the slopes of entire surface | $\theta_{2 s}=\sqrt{\frac{\sum \sum \tan ^{2} \alpha_{i j}}{n}}$ | $Z_{2 s} ; S_{d q} ; Z_{2}$ |  |  | Belem et al. (2000); Zhang et al. (2009) |
| $\theta_{2 s i}$ | Root mean square of the slopes facing shearing direction | $\theta_{2 s i}=\sqrt{\frac{\sum \sum \tan ^{2} \alpha_{i j s}}{m_{s}}}$ | $Z_{2, k}$ | $\checkmark$ | $\checkmark$ | Zhang et al. (2009) | $\alpha_{i j}$ actual dip angle of the elements of 3D wireframe, $\alpha_{i j, s}$ actual dip angle of elements facing shearing direction of 3D wireframe, $\theta^{*}{ }_{\text {max }}$ maximum apparent angle of the triangular elements in shearing direction, 3 D three-dimensional, Anisotropy parameters allowing consideration of shear direction, $C$ a fitted value, $\mathrm{E}^{J R C}$ parameters used in published empirical equations, $m_{s}$ total number of the elements facing shearing direction of 3D wireframe, $n$ total number of the elements of 3D wireframe

Table 2 Three-dimensional area parameters describing fracture surface roughness

| Term | Definition | Calculation | Original | Anisotropy | $\mathrm{E}^{\text {JRC }}$ |
| :--- | :--- | :--- | :--- | :--- | :--- | References.

$A_{t}$ actual area, $A_{b}$ : brightened area of fracture surface under a simulated parallel light, $A_{n}$ nominal area, $\cos \phi$ tortuosity index of the least square plane of the four extreme marginal points of the surface
surface. Based on this, Grasselli (2001) proposed a 3D parameter $\theta^{*}{ }_{\text {max }} / C$ :
$A_{\theta^{*}}=A_{0}\left(\frac{\theta^{*}{ }_{\max }-\theta^{*}}{\theta^{*}{ }_{\max }}\right)^{C}$
where $\theta^{*}{ }_{\text {max }}$ is the maximum apparent dip angle of the elements against shearing direction, $C$ is a fitted value calculated via nonlinear least-squares regression, and $\mathrm{A}_{0}$ is the value of $A_{\theta^{*}}$ when $\theta^{*}$ equals 0 . However, Cottrell (2009) argued that parameter suggested by Grasselli (2001) has no physical meaning when $C$ equals 0 and revised it by proposing parameter $\theta_{\text {max }}^{*} /(1+C)$, which is accepted by Tatone and Grasselli (2010). $\theta^{*}{ }_{\max }(1+C)$ is abbreviated as $\theta_{c i}$ for later use in the present study.

Belem et al. (2000) proposed the mean $\left(\theta_{s}\right)$ and the root mean square $\left(\theta_{2 s}\right)$ of the actual dip angle of the elements for an entire surface to describe surface roughness. Lee et al. (2011) updated $\theta_{s}$ as proposed by Belem et al. (2000) to $\theta_{s i}$ by adopting the elements facing the shear direction. Zhang et al. (2009) revised $\theta_{2 s}$ as proposed by Belem et al.(2000) into $\theta_{2 s i}$ by adopting only the elements facing the shear direction. Definitions and calculations of the slope parameters are shown in Table 1.

Parameters related to area include actual area, nominal area, shearing area, etc. of the elements or entire surface. The ratio $\left(S_{s}\right)$ between the actual area and the nominal area was first defined as a surface roughness index by El-Soudani (1978). $S_{s}$ was revised by Belem et al. (2000) for both upper and lower fracture surfaces. Grasselli (2001) stated that it is correct

Table 3 Three-dimensional volume and amplitude parameters describing fracture surface roughness

| Term | Definition | Parameter | Original | References |
| :---: | :---: | :---: | :---: | :---: |
| $Z_{\text {rms }}$ | Root mean square height of z | $Z_{r m s}=\left(\frac{1}{N} \iint z_{i j}^{2} d x d y\right)^{1 / 2}$ | $R M S ; S_{q}$ | Marlinverno (1990) Fan et al. (2013) |
| $Z_{S z}$ | Ten-point height | $Z_{S z}=\frac{\sum_{1}^{5}\left\|z_{p i}\right\|+\sum_{1}^{5}\left\|z_{z}\right\|}{10}$ | $S_{z}$ | ISO 25178-2: 2012 |
| $Z_{s a}$ | Mean of the absolute of z | $Z_{s a}=\frac{1}{m \times n} \sum_{j=1}^{m} \sum_{I=1}^{n}\left\|z_{i j}\right\|$ | $S_{a}$ | ISO 25178-2: 2012 Fan et al. (2013) |
| $Z_{s s}$ | Skewness of the surface | $Z_{s s}=\frac{\frac{1}{m \times n} \sum_{j=1}^{m} \sum_{i=1}^{n} z_{i j}{ }^{3}}{Z_{m s s^{3}}^{3}}$ | $S_{s k}$ | ISO 25178-2: 2012 |
| $Z_{\text {sk }}$ | Kurtosis of the surface | $Z_{s k}=\frac{\frac{1}{m \times n} \sum_{i=1}^{m} \sum_{i=1}^{n} z_{i j}{ }^{4}}{Z_{m s}{ }^{4}}$ | $S_{k w}$ | ISO 25178-2: 2012 |
| $Z_{\text {range }}$ | Range of z | $Z_{\text {range }}=z_{\text {max }}-z_{\text {min }}$ | Range; $S_{p}$ | ISO 25178-2: 2012 Fan et al. (2013) |
| $V_{s v i}$ | Valley fluid retention index | $V_{s v i}=\frac{V_{v}\left(h_{0.8}\right)}{Z_{m m s}(M-1)(N-1) \Delta x \Delta y}$ | $S_{v i}$ | ISO 25178-2: 2012 |
| $V_{s c i}$ | Core fluid retention index | $V_{s c i}=\frac{V_{v}\left(h_{0.05}\right)-V_{v}\left(h_{0.8}\right)}{Z_{m m s}(M-1)(N-1) \Delta x \Delta y}$ | $S_{c i}$ | ISO 25178-2: 2012 |

$V_{v}\left(h_{0.8}\right)$ void volume at the surface heights at $80 \%$ bearing area, $V_{v}\left(h_{0.05}\right)$ void volume at the surface heights at $5 \%$ bearing area, $Z_{i j}$ heights of point cloud of scanned surface, $z_{p i}$ five highest peaks of point cloud of scanned surface $(i=1,2, \ldots, 5), z_{v i}$ five lowest troughs of point cloud of scanned surface, $Z_{\text {max }}$ highest peaks of point cloud of scanned surface, $Z_{\text {min }}$ lowest troughs of point cloud of scanned surface


Fig. 1 Tested rock joint samples: (a) lower fracture surface; and (b) upper fracture surface. The dots show the pattern of points tested for strength of the joint wall (JCS). D1-D4 indicate the four succeeding shearing directions in direct shear tests on each specimen
to assume that the upper and lower fracture surfaces of fresh joint are in $100 \%$ contact for JRC estimation.

Ge et al. (2012) proposed $S_{\text {bap }}$, the percentage of the bright area over the actual fracture surface area, as a 3 D roughness parameter. However, they did not specify an optimum incident angle for the parallel light for the estimation of JRC. Belem (2000) proposed similar 3D roughness parameters, $S_{d r}$ and $S_{t s}$, which are also based on $A_{t}$ and $A_{n}$ (nominal area). Details of area parameters are listed in Table 2.

Most parameters related to amplitude or volume of the surface elements are cited by geometrical specifications (ISO 25178-2: 2012). There are six parameters $\left(Z_{r m s}, Z_{s z}\right.$, $Z_{s a}, Z_{s s}, Z_{s k}$, and $Z_{\text {range }}$ ) related to amplitude and two parameters ( $V_{s v i}$ and $V_{s c i}$ ) related to volume. Definitions and calculations of these parameters are given in Table 3.

Although many 3D parameters were suggested for quantifying surface roughness, only a few slope and area parameters were examined in developing empirical equations predicting $\operatorname{JRC}\left(\theta_{c i}, \theta_{s}, \theta_{s i}, \theta_{2 s i}\right.$, and $S_{b a p}$ in Tables 1 and 2). Grasselli's (2001) experiments and corresponding roughness parameters $\left(\theta_{c i}\right)$ have no direct relationship with the JRC of rock fracture surfaces. Empirical equations proposed by Tatone (2009), which are linked to JRC, are only based on the ten 2D standard profiles by Barton (1976). Zhang et al. (2009) used the 3D parameter $\theta_{2 s i}$ in a 2 D empirical equation by Tse and Cruden (1979). Ge et al. (2012) derived an empirical equation for an unknown incident angle of the parallel light. In addition, few 3D empirical equations take SIs into consideration. Lee et al. (2011) made a comparison of JRC with surface angularity $\theta_{s}$ at SIs of $0.2,0.3,0.5,1.0,2.0$, and 5.0 mm but suggested empirical equations with a fixed optimum sampling interval. However, it has been stated by Yang et al. (2001), Li et al. (2016), Zheng and Qi (2016), and Liu et al. (2017) that sampling intervals might shift the relationship between the JRC and roughness parameters.

Considering the restrictions and limited scope of the abovementioned findings, a set of reliable 3D roughness parameters and correlations between such parameters and JRC are required. This objective can be successfully achieved via direct shear tests

Table 4 Mechanical properties of the fracture surfaces

| Sample no. | $\varphi_{\mathrm{b}}\left({ }^{\circ}{ }^{\text {a }}\right.$ | JCS (MPa) | $\sigma_{\mathrm{n}} / \mathrm{JCS}$ | $\tau(\mathrm{kPa})$ | STD |
| :---: | :---: | :---: | :---: | :---: | :---: |
| 1 | 28.0 | 33.62 | 0.01 | 585.38 | 60.1 |
| 2 | 30.6 | 25.93 | 0.02 | 672.42 | 35.2 |
| 3 | 28.0 | 15.02 | 0.03 | 710.05 | 135.0 |
| 4 | 28.0 | 28.69 | 0.02 | 670.76 | 83.5 |
| 5 | 28.0 | 26.41 | 0.02 | 458.71 | 48.3 |
| 6 | 28.0 | 31.94 | 0.02 | 629.93 | 160.2 |
| 7 | 28.0 | 21.95 | 0.02 | 617.60 | 91.1 |
| 8 | 28.0 | 20.13 | 0.02 | 619.29 | 143.8 |
| 9 | 28.0 | 18.30 | 0.03 | 356.84 | 93.3 |
| 10 | 28.0 | 28.95 | 0.02 | 607.11 | 47.0 |
| 11 | 29.7 | 36.06 | 0.01 | 352.57 | 16.7 |
| 12 | 28.0 | 23.08 | 0.02 | 703.71 | 66.0 |
| 13 | 24.1 | 71.19 | 0.01 | 223.50 | 8.9 |
| 14 | 27.9 | 39.29 | 0.01 | 530.50 | 128.5 |
| 15 | 27.9 | 42.99 | 0.01 | 697.75 | 50.0 |
| 16 | 27.9 | 41.50 | 0.01 | 659.50 | 97.7 |
| 17 | 27.9 | 37.49 | 0.01 | 627.00 | 66.5 |
| 18 | 27.9 | 36.27 | 0.01 | 717.75 | 125.5 |
| 19 | 27.9 | 40.94 | 0.01 | 641.50 | 93.0 |
| 20 | 27.9 | 33.71 | 0.01 | 820.75 | 61.2 |
| 21 | 27.9 | 40.71 | 0.01 | 668.50 | 73.8 |
| 22 | 27.8 | 53.76 | 0.01 | 852.79 | 65.8 |
| 23 | 27.9 | 27.37 | 0.02 | 248.50 | 61.2 |
| 24 | 27.9 | 25.47 | 0.02 | 453.25 | 91.4 |
| 25 | 27.9 | 71.01 | 0.01 | 264.75 | 4.8 |
| 26 | 27.8 | 47.47 | 0.01 | 491.19 | 80.7 |
| 27 | 27.9 | 31.85 | 0.02 | 695.75 | 73.4 |
| 28 | 27.8 | 42.75 | 0.01 | 460.87 | 47.8 |
| 29 | 27.8 | 51.81 | 0.01 | 537.75 | 58.2 |
| 30 | 27.8 | 35.76 | 0.01 | 706.17 | 72.8 |
| 31 | 27.8 | 47.09 | 0.01 | 518.86 | 47.4 |
| 32 | 27.8 | 67.60 | 0.01 | 406.13 | 73.3 |
| 33 | 27.8 | 67.60 | 0.01 | 409.09 | 26.3 |
| 34 | 27.8 | 51.81 | 0.01 | 480.37 | 97.9 |
| 35 | 27.8 | 63.20 | 0.01 | 473.90 | 66.5 |
| 36 | 27.8 | 56.50 | 0.01 | 417.60 | 77.3 |
| 37 | 27.8 | 55.62 | 0.01 | 448.68 | 39.6 |
| 38 | 27.9 | 44.76 | 0.01 | 645.50 | 98.8 |

$J C S$ strength of the joint wall, STD standard deviation of measured peak shear strengths
on a large sample population and success in achieving 3D roughness parameters.

## Experiments

Mechanical properties and surface geometry of rock joints are the two key parts of this study. Sample preparations,

Fig. 2 Representative plots of direct shear tests: (a) shear stress vs. shear displacement; and (b) vertical displacement vs. shear displacement. D1-D4 indicate the four succeeding shearing directions in direct shear tests on each specimen

b


Fig. 3 Flowchart for determining the three-dimensional roughness parameters of a rock joint. 3D three-dimensional, TIN Triangulated Irregular Network

mechanical tests, and surface measurements to describe these components are summarized in the following sections.

## Sample preparation

We collected 38 groups of fresh rock blocks with structural planes in the middle, of which nine were limestone, 12 granite, and 17 sandstone. The structural planes of these samples have varied roughness, from smooth to extremely rough, forming a sequence. Samples of limestone and sandstone were collected from rock cores. The rest were artificially produced by splitting granite blocks. The upper and lower fracture surfaces of all specimens were fresh and matched well, showing no obvious aperture and infilling. The length and width of the rock specimens were restricted to $120 \mathrm{~mm} \times 120 \mathrm{~mm}$ and each specimen
was cemented into a $150 \mathrm{~mm} \times 150 \mathrm{~mm} \times 223.5 \mathrm{~mm}$ concrete block. Great attention was paid to aligning the joint surfaces of the concreted specimens so that they were as horizontal as possible (Fig. 1). This was done by putting reference lines, which were parallel to the main inclination of the joint surface, along the periphery of the specimens and rotating the specimens to make the reference lines aligned horizontally during sample preparation.

## Mechanical properties

According to Barton's JRC-JCS model, the JRC of rock fracture can be back-calculated if the strength of peak shear strength $(\tau)$, joint wall (JCS), and basic friction angle $\left(\varphi_{b}\right)$ of the fracture were known.

Fig. 4 The morphological reconstruction of the fracture surface: (a) Triangulated Irregular Network (TIN); and (b) threedimensional wireframe

b

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Peak shear strength of the 38 prepared joint specimens were measured using the direct shear test. The normal stress was set to 500 kPa , corresponding a depth of $15-25 \mathrm{~m}$, to simulate the actual in situ conditions of the samples. The shear rate was $0.3 \mathrm{~mm} / \mathrm{min}$ for all specimens. Each specimen's four sides are marked as D1-D4 and sheared in these directions successively (Fig. 1). The average value and standard deviation of measured peak shear strengths (D1-D4) of each specimen are given in Table 4. Representative curves of shear stress and vertical displacement versus shear displacement are shown in Fig. 2. The curves of shear stress versus shear displacement follow the similar pattern for all four directions.

In the determination of the rebound value, 32 points on each specimen are tested using a Schmidt rebound hammer (Fig. 1). The rebound value for each surface is calculated from the average. The JCS of the specimen is then calculated according to Aydin (2009). The basic-friction angle $\left(\varphi_{b}\right)$ is measured using tilt tests on saw-cut dry joint surfaces. The mechanical properties of the fracture surfaces are listed in Table 4.

## Surface measurements

With the help of laser scanning technology (HAND SCAN ${ }^{\text {тм }}$ 300 , resolution of 0.1 mm and accuracy of 0.04 mm ), the morphology of each fracture surface can be digitized into a point cloud. In this study, both upper and lower fracture surfaces of the 38 sets of samples were scanned before and after shearing with 0.2 mm spatial resolution. Fig. 3 shows the steps to derive roughness parameters of joint surfaces. Grid data sets were constructed by sampling the point cloud at different intervals $(0.4,0.8,1.6,3.2$, and 6.4 mm$)$ using Microsoft Excel ${ }^{\circledR}$. The grid data sets were then used to build TIN and 3D wireframe models (Fig. 4) using ArcGIS ${ }^{\circledR}$, Surfer ${ }^{\circledR}$, and MATLAB ${ }^{\circledR}$ programs. The amplitude parameters (e.g., $Z_{s a}, Z_{\mathrm{rms}}$ and $Z_{\text {range }}$ ) were derived directly from the 3D point cloud through Excel ${ }^{\circledR}$ calculation, slope parameters $\left(\theta_{s}\right.$ and $\left.\theta_{2 s}\right)$ were derived from TIN models, volume parameters ( $V_{a n}$ ) were derived from 3D wireframe models, and area parameters ( $S_{s T}$ and $S_{S F}$ ) were derived from TIN and 3D wireframe models using Excel ${ }^{\circledR}$.

The variations ( $V$ ) of amplitude parameters $\left(Z_{r m s}, Z_{s a}\right.$, $Z_{\text {range }}$ ) between upper and lower surfaces (e.g., $V_{r m s}=$ the ratio of [ $Z_{r m s}$ of upper surface $-Z_{r m s}$ of lower surface] to $Z_{r m s}$ of lower surface) were examined and are shown in Table 5. It was found that the maximum variation is about $-4.68 \%$, indicating that both sides of the joints match well. Grasselli (2001) and Belem et al. (2000) demonstrated that fresh joints have few voids in between both sides. Fan et al. (2013) also indicated that the $S_{s}$ and $\theta_{2 s}$ values of each set of coupled joints are quite close.

The first three shear tests (D1, D2, and D3) ceased at a shear displacement of about 6 mm to capture the peak strength. The last shearing (D4) was stopped at a shear displacement of more than 30 mm . This strategy was originally
designed to gain the residual strength of the joint from the last test. The difference of amplitude parameters between pre- and post-shear fracture surfaces is less than $2.16 \%$ (Table 5), indicating there is no obvious damage induced by shearing. This is

Table 5 Variation of height parameters between upper and lower joint surfaces and difference of height parameters between pre- and post-shear surfaces

| Sample no. | Variation of height parameters between upper and lower surface (\%) |  |  | Difference of height parameters between preand post-shear surfaces (\%) |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  | $V_{r m s}$ | $V_{s a}$ | $V_{\text {range }}$ | $D_{\text {rms }}$ | $D_{s a}$ | $D_{\text {range }}$ |
| 1 | - 2.78 | - 3.18 | - 3.49 | 0.30 | 0.29 | 0.44 |
| 2 | 1.41 | 1.48 | 1.49 | 0.08 | 0.09 | 0.13 |
| 3 | - 2.43 | -2.61 | - 2.70 | 0.29 | 0.30 | 0.4 |
| 4 | 1.28 | 1.44 | 1.50 | 0.90 | 0.90 | 1.35 |
| 5 | 2.89 | 2.99 | 3.10 | 0.41 | 0.42 | 0.42 |
| 6 | - 0.91 | - 1.02 | - 1.08 | 1.14 | 1.17 | 1.18 |
| 7 | 2.73 | 3.15 | 3.20 | 0.04 | 0.05 | 0.05 |
| 8 | 2.84 | 3.20 | 3.51 | 0.24 | 0.25 | 0.25 |
| 9 | - 3.07 | -3.52 | - 3.83 | 0.37 | 0.37 | 0.56 |
| 10 | - 3.19 | - 3.72 | - 3.76 | 1.24 | 1.27 | 1.28 |
| 11 | 1.63 | 1.76 | 1.76 | 1.10 | 1.12 | 1.13 |
| 12 | 2.50 | 2.75 | 2.90 | 0.19 | 0.19 | 0.28 |
| 13 | 2.47 | 2.76 | 2.78 | 0.71 | 0.72 | 1.08 |
| 14 | - 4.68 | -4.68 | - 2.93 | 0.14 | 0.14 | 0.23 |
| 15 | 1.37 | 1.37 | 1.62 | 0.06 | 0.06 | 0.26 |
| 16 | - 3.02 | - 3.01 | 0.83 | 0.06 | 0.06 | 1.49 |
| 17 | - 2.25 | - 2.25 | - 0.49 | 0.10 | 0.11 | 1.43 |
| 18 | 0.56 | 0.59 | 0.30 | 0.32 | 0.32 | 1.52 |
| 19 | 2.54 | 2.60 | 2.62 | 0.60 | 0.60 | 0.10 |
| 20 | 1.19 | 1.32 | 1.45 | 0.76 | 0.76 | 1.44 |
| 21 | - 0.18 | - 0.20 | - 0.21 | 0.24 | 0.24 | 0.61 |
| 22 | 2.34 | 2.81 | 3.04 | 0.29 | 0.29 | 0.44 |
| 23 | 1.84 | 2.03 | 2.09 | 1.09 | 1.07 | 1.60 |
| 24 | 0.96 | 1.05 | 1.10 | 0.87 | 0.89 | 0.90 |
| 25 | 0.48 | 0.56 | 0.59 | 1.07 | 0.64 | 1.34 |
| 26 | 2.40 | 2.79 | 2.89 | 0.66 | 0.65 | 1.23 |
| 27 | - 2.12 | - 2.33 | - 2.33 | 0.47 | 0.48 | 2.16 |
| 28 | 2.99 | 3.38 | 3.65 | 1.04 | 1.06 | 2.02 |
| 29 | - 1.60 | - 1.78 | - 1.93 | 0.89 | 0.87 | 1.31 |
| 30 | - 2.51 | - 2.77 | - 2.87 | 1.15 | 1.17 | 1.18 |
| 31 | 1.65 | 1.87 | 2.03 | 0.09 | 0.09 | 0.09 |
| 32 | 2.34 | 2.51 | 2.73 | 1.14 | 1.17 | 1.18 |
| 33 | 2.77 | 3.02 | 3.05 | 0.66 | 0.66 | 0.67 |
| 34 | 0.77 | 0.86 | 0.93 | 0.52 | 0.53 | 0.54 |
| 35 | - 3.13 | - 3.39 | - 3.55 | 0.12 | 0.13 | 0.13 |
| 36 | 2.82 | 2.89 | 3.06 | 1.00 | 1.02 | 1.03 |
| 37 | 0.73 | 0.81 | 0.84 | 0.45 | 0.46 | 0.47 |
| 38 | - 1.57 | - 1.79 | - 1.81 | 0.28 | 0.27 | 0.36 |

$D_{\text {range }}$ difference of $Z_{\text {range }}\left(D_{\text {range }}=\frac{Z_{\text {range }}{ }^{p o s t}-Z_{\text {range }}^{p r e}}{Z_{\text {range }}^{\text {pre }}}\right.$ ), $D_{\text {rms }}$ difference of



the superscript "upper" and "lower" stand for the parameter derived from the upper or lower joint surface, respectively; the superscript "post" indicates that the parameter is derived from joint surface after the 4th shearing, while "pre" means that the parameter is derived from joint surface before the 1st shearing


Fig. 5 Photos of fracture surface of sample 1 sheared in different directions of: (a) D1; (b) D2; (c) D3; and (d) D4. D1-D4 indicate the four succeeding shearing directions in direct shear tests on each specimen
evident in the photos of post-shear joint surfaces (Fig. 5). The stress level ( $500 \mathrm{kPa} / \mathrm{JCS}$ ) is less than 0.03 for all tests (Table 4). This also protects the fracture surface from being damaged during shearing. Considering this, the lower surface was used to represent the coupled joint surfaces and its original morphology was used to generate the 3D roughness parameters for later calculation and analysis.

## Empirical equations for the entire surface

Considering the anisotropy of rock joints and uncertainty of shear direction in real rock engineering, we propose use of average peak strength (average value from the four direction shears) for the back-calculation of JRC. Accordingly, the roughness parameters should be direction-independent and valid for the entire surface.

## Slope parameters

As shown in Table 1, $\theta_{s}$ and $\theta_{2 s}$ do not consider shear direction, whereas $\theta_{s i}, \theta_{2 s i}$, and $\theta_{c i}$ do. Therefore, $\theta_{s}$ and $\theta_{2 s}$ were chosen and calculated for deriving empirical equations with the backcalculated JRC. In calculation of $\theta_{s}$ and $\theta_{2 s}$, previous researchers suggested deriving $\alpha_{i j}$ (dip angle of elements of the entire surface) from the 3D wireframe model (Belem et al. 2000, Lee et al. 2011 and Zhang et al. 2009). However, not every element in the 3D wireframe model is a quadrangle with all four corners in the same plane. This makes calculating $\alpha_{i j}$ difficult and inaccurate, since a least-square plane has to be constructed to substitute the real element. We suggest use of the TIN model rather than the 3D wireframe model to calculate $\alpha_{i j}$ and then slope parameters to reduce the calculation-induced deviation.

On the other hand, as $\theta_{c i}$ is also a parameter considering shear direction, we propose an equivalent parameter $\theta_{g}$ (Eq. 3 ), which is the integral of $A_{\alpha} . A_{\alpha}$ is the ratio of $A_{p}$ to $A_{t}$, where $A_{p}$ is the total area of triangular elements whose actual dip angle is larger than $\alpha$ and $A_{t}$ is the actual area of the surface.
$\theta_{g}=\int_{0}^{a_{\text {max }}} A_{\alpha} d \alpha$
where, $\alpha_{\max }$ is the maximum actual dip angle of the triangular elements.
$\theta_{g}$ avoids using the fitted value $C$ (for calculating $\theta_{c i}$ ), and it is much easier to calculate. The newly derived JRC estimation equations as functions of $\theta_{s}, \theta_{2 s}$, and $\theta_{g}$ are listed in Table 6 and plotted in Fig. 6.

## Area parameters

According to the definition of $S_{s}$ in Table 2, two parameters, $S_{S F}$ and $S_{S T}$, were obtained for each fracture surface from the constructed 3D wireframe and TIN models, respectively. Both of them show close correlations with JRC as shown in Table 6 and Fig. 7.
$S_{d r}$ and $S_{s}$ have a similar physical meaning (Table 2). In this study, $S_{d r}$ exhibits lower correlation coefficients when it is

Table 6 Empirical equations of selected joint roughness coefficient coupling roughness parameters with sampling interval as independent variables

| Variable | Equation | $\mathrm{R}^{2}$ |
| :--- | :--- | :--- |
| $\theta_{s}$ | $J R C=2.9 \theta_{s}^{0.7}+S I^{0.809}-5.6$ | 0.902 |
| $\theta_{g}$ | $J R C=3.9 \theta_{g}{ }^{0.6}+S I^{0.815}-7.7$ | 0.900 |
| $\theta_{2 s}$ | $J R C=36.5 S_{2 s}^{0.5}+S I^{0.845}-7.8$ | 0.883 |
| $S_{s T}$ | $J R C=47.7\left(S_{s T}-1\right)^{0.3}+S I^{0.836}-6.7$ | 0.889 |
| $S_{s F}$ | $J R C=47.8\left(S_{S F}-1\right)^{0.3}+S I^{0.837}-6.7$ | 0.888 |
| $V_{a n}$ | $J R C=43.9 V_{a n}^{0.1}+S I^{0.087}-34.4$ | 0.824 |
| $Z_{s a}$ | $J R C=58.5 Z_{s a}{ }^{0.1}-S I^{0.071}-48.9$ | 0.804 |
| $Z_{\text {rms }}$ | $J R C=63.2 Z_{\text {rms }}^{0.1}+S I^{0.090}-54.7$ | 0.798 |
| $Z_{\text {range }}$ | $J R C=15.5 Z_{\text {range }}{ }^{0.2}+S I^{0.166}-15.8$ | 0.759 |

[^0]Fig. 6 Multiple regression of joint roughness coefficient (JRC), sampling interval (SI), and roughness parameters of (a) $\theta_{s}$; (b) $\theta_{g}$; and (c) $\theta_{2 s}$


used to get regression correlations with JRC than $S_{S F}$ and $S_{S T}$. Regression analysis was also done for $S_{t s}$, which gives very low correlation coefficient. This may be due to insufficient consideration by taking only four corner points of the fracture surface to calculate the tortuosity index $(\cos \phi)$ of the entire surface. We therefore excluded using $S_{d r}$ and $S_{t s}$ for the estimation of JRC.

## Volume and amplitude parameters

Amplitude parameters $Z_{s a}$ and $Z_{s z}$ are based on the same measurements and are closely related. In this study, $Z_{s a}, Z_{r m s}, Z_{s s}$, $Z_{s k}$, and $Z_{\text {range }}$ were used. In addition, $V_{s v i}$ and $V_{s c i}$ demonstrated no good correlation with JRC (the correlation coefficients
are less than 0.4 ). We propose a parameter, $V_{a n}$, which is the ratio of the net volume of fracture surface $\left(V_{n}\right)$ to the projected area $\left(A_{n}\right)$. The net volume $\left(V_{n}\right)$ is the summation of positive and negative volumes of the surface segmented by the least-square plane (Fig. 8). It is found that $V_{a n}$ has close correlation with JRC. Table 6 lists the newly derived equations for volume and amplitude parameters, which are plotted in Fig. 9. The parameters $Z_{s k}$ and $Z_{s s}$ show no relation with JRC.

## Discussion

In general, the nine proposed equations in Table 6 are all capable of estimating the JRC of rock fracture surfaces as they

Fig. 7 Multiple regression of joint roughness coefficient (JRC), sampling interval (SI), and roughness parameters of (a) $S_{S T}$; and (b) $S_{S F}$



Fig. 8 Measurement of net volume $\left(V_{n}\right)$ of the fracture surface

have correlation coefficients greater than 0.75 . Among them, slope parameters perform the best and the amplitude parameters perform the worst in terms of correlation coefficient. Regarding the usability and applicability, the amplitude parameters ( $Z_{s a}, Z_{\mathrm{rms}}$, and $Z_{\text {range }}$ ) can be directly and easily calculated in Excel ${ }^{\circledR}$ once the point cloud is obtained by scanning the fracture surface. The calculation of $V_{a n}$ and $S_{S F}$ is based on 3D wireframe and that of slope parameters $\left(\theta_{s}, \theta_{g}\right.$, and $\theta_{2 s}$ ) and $S_{S T}$ is based on the TIN model. They all require third-party software programs (e.g., Surfer ${ }^{\circledR}$, MATLAB ${ }^{\circledR}$, and ArcGIS ${ }^{\circledR}$ in this study) to deal with the point could. Considering the difficulties in a complex calculation for slope, area and volume parameters, one can chose the amplitude parameters to facilitate rapid estimation of JRC in engineering practice.

The regression correlations in Table 6 indicate that the sampling interval has some influence on the estimation of JRC, especially when slope and area parameters are used. Li et al. (2016) and Liu et al. (2017) also found that slope parameters $\left(Z_{2}, \beta_{100 \%}\right.$, and $\left.\sigma_{i}\right)$ show much higher sensitivity to the sampling interval than amplitude parameters $\left(R_{z}, \lambda\right.$, and $\left.D_{h-L}\right)$
in the determination of 2D JRC. Although the sampling intervals used in this study ( $0.2-6.4 \mathrm{~mm}$ ) cover a wide scope, great caution should be paid when employing the proposed equations for other sampling intervals.

The stress level ( $500 \mathrm{kPa} / \mathrm{JCS}$ ) used in this study was designed for simulating the actual in situ conditions of the tested samples, which were collected from the depth of $15-25 \mathrm{~m}$, and for protecting the fracture surface from being damaged during shearing. The proposed equations are suggested to be used for rock joints in shallow layers or for joints whose surfaces are hardly altered during shearing.

## Conclusion

For decades, objective and quantitative determination of JRC were investigated mostly for the parameters derived from a 2D profile. However, a single or multiple straight-line profiles collected from a fracture surface cannot reflect the roughness of the entire surface. This study reviews roughness parameters

Fig. 9 Multiple regression of joint roughness coefficient (JRC), sampling interval (SI), and roughness parameters of (a) $V_{a n}$; (b) $Z_{s a}$; (c) $Z_{r m s}$; and (d) $Z_{\text {range }}$

derived from 3D surfaces and conducts relevant experiments. Back-calculated JRC values from 38 rock blocks with existing fractures are used to derive new empirical equations as a joint function of roughness parameter and sampling interval for JRC estimation.

The following main conclusions can be made:
(1) Nine parameters $\left(\theta_{s}, \theta_{g}, \theta_{2 s}, S_{s T}, S_{s F}, V_{a n}, Z_{s a}, Z_{\mathrm{rms}}\right.$, and $Z_{\text {range }}$ ) are found to have close correlations with JRC and are capable of estimating the JRC of rock fracture surfaces. Other parameters $\left(Z_{s s}, Z_{s k}, V_{s v i}, V_{s c i}, S_{d r}\right.$, and $\left.S_{t s}\right)$ show no good correlations with JRC.
(2) Slope parameters perform the best and the amplitude parameters perform the worst in terms of correlation coefficient.
(3) The sampling interval has little influence when using volume and amplitude parameters ( $V_{\mathrm{an}}, Z_{s a}, Z_{r m s}$, and $Z_{\text {range }}$ ), while it influences to some extent when other parameters $\left(\theta_{s}, \theta_{g}, \theta_{2 s}, S_{s T}\right.$, and $\left.S_{s F}\right)$ are used.

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[^0]:    $\theta_{g}$ threshold angle $\left(\theta_{g}=\int_{0}^{\alpha_{\text {max }}} A_{\alpha} d \alpha\right), A_{\alpha}$ threshold ratio for the entire surface ( $A_{\alpha}=A_{p} / A_{n}$ ), $A_{p}$ total area of the triangular elements whose actual dip angles are larger than $\alpha, A_{n}$ nominal area, JRC joint roughness coefficient, $S_{S F}$ ratio of actual and nominal area of the three-dimensional wireframe, $S I$ sampling interval, $S_{S T}$ ratio of actual and nominal area of the Triangulated Irregular Network (TIN) surface, $V_{a n}$ ratio of the net volume of fracture surface to the projected area $\left(V_{a n}=V_{n} / A_{n}\right), V_{n}$ the summation of positive and negative volumes of the surface segmented by the least-square plane

