**ORIGINAL PAPER**



# **Determination of meaningful block sizes for rockfall modelling**

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#### **Abstract**

The determination of the so-called design block is one of the central elements of the Austrian guideline for rockfall protection ONR 24810. It is specifed as a certain percentile (P95–P98, depending on the event frequency) of a recorded block size distribution. Block size distributions may be determined from the detachment area (in situ block size distribution) and/or from the deposition area (rockfall block size distribution). Deposition areas, if present, are generally accessible and measurable without technical aids. However, most measuring methods are subjective, uncertain, not verifable, or inaccurate. Also, rockfall blocks are often fragmented due to the preceding fall process. The in situ block size distribution is (also) required for meaningful rockfall modelling. The statistical method seems to be the most efficient and cost-effective method to determine in situ block size distributions with many blocks within the whole range of block sizes. In the current literature, joint properties are often described by the lognormal and exponential distribution functions. Today, we can model synthetic rock masses on the basis of discrete fracture networks. They statistically describe the geometric properties of the joint sets. This way, we can carry out exact rock mass block surveys and determine in situ block size distributions. We wanted to know whether the in situ block size distributions derived from the synthetic rock mass models can be described by probability distribution functions, and if so, how well. We ftted various distribution functions to three determined in situ block size distributions of diferent lithologies. We compared their correlations using the Kolmogorov–Smirnov test and the mean-squared error method. We show that the generalized exponential distribution function best describes the in situ block size distributions across various lithologies compared to 78 other distribution functions. This could lead to more certain, accurate, verifable, holistic, and objective results. Further investigations are required.

**Keywords** Rock fall · Block size distribution · Design block · Joint spacing · Persistence · Joint size · Discrete joint network · Synthetic rock mass model · Probability distribution

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## **1 Introduction**

The Austrian Guideline ONR 24810 (On [2021](#page-25-0)) regulates the design of rockfall protection nets, embankments, and galleries. The design is based on the kinetic energy and bounce height of a so-called design block. The design block is specifed as a percentile (P95-P98) of a recorded block size distribution (BSD), depending on the event frequency. BSDs may be determined from the detachment area (in the following referred to as in situ block size distribution—IBSD) and/or from the deposition area (in the following referred to as rockfall block size distribution—RBSD). The ONR 24810 proposes various methods to determine BSDs. In the detachment area, these include estimation by visual assessment, statistical IBSD and discrete explicit block measurement. In the deposition area, these include estimation, random axis measurements, the line-counting method, the area method, sieve analysis or photosieving by software (Gaich and Pötsch [2022\)](#page-25-1).

Deposition areas, if available, are generally accessible and measurable without technical aids. However, most measuring methods are subjective, uncertain, not verifable, or inaccurate. Blocks smaller than a fist are usually not measured. Often, (very) small block volumes are underrepresented, unnoticed or considered minor events and not included in rockfall inventories (De Biagi et al. [2017](#page-25-2); Laimer [2019](#page-25-3)). This affects the percentiles of the BSD. Neglecting many small blocks in a BSD results in bigger P95–P98 blocks. In hazard analyses, smaller blocks may play an important role, depending on the protection target. For most realistic distributions of kinetic energy, bounce height and runout, the entire BSD should be considered in rockfall models (Illeditsch and Preh [2020](#page-25-4)). Also, rockfall blocks are often fragmented due to the preceding fall process. RBSDs determined from a few dozen blocks (or defined design blocks thereof) do not seem to be sufficient for meaningful rockfall modelling. If possible, IBSDs should (also) be considered. Visual assessment of rock faces seems subjective and not verifable. Discrete explicit block measurements with rope access require intensive resources and time and are often unfeasible. The statistical method seems the most efficient and cost-effective method to determine IBSDs with many blocks within the full range of block sizes.

The distribution of in situ block sizes (IBSDs) is based on the knowledge that bedding, foliation and joints are commonly closely spaced and have low persistence. Wide spacing and high persistence joints are less common. Consequently, for rock blocks formed by joints, small blocks are more common than large blocks (Wyllie [2014](#page-25-5)). In a case study, Wyllie ([2014\)](#page-25-5), Chapter 8.3.1, estimates the means and standard deviations of lognormal distributions for the rockfall diameter [m] and thickness [m] of discoid-shaped blocks. Based on these distributions, he determines an IBSD  $[m<sup>3</sup>]$ . Priest and Hudson [\(1981](#page-25-6)) show a histogram of measured joint sizes [m] (i.e., persistence) in a Cambrian sandstone. They fit both exponential and lognormal curves to this data, for which the correlation coefficients *r* are 0.69 and 0.89, respectively. While the lognormal curve has a higher correlation coefficient, the exponential curve has a better fit at longer joint sizes. Hudson and Harrison  $(1997)$  $(1997)$ , Chapter 7.2.1, show that, when a sufficient large sample of individual joint spacings [m] (more than 200) is plotted in histogram form, a negative exponential distribution is often evident. Palmström [\(2000](#page-25-8)) measures the orientation and spacing of three joint sets on a horizontal and vertical surface. He cuts a cube of 10 m edge length by these joint sets and plots the distribution curve of the resulting blocks  $[m<sup>3</sup>]$  in logarithmic scale. The s-shaped curve reminds of a sieve curve. He suggests characterizing the curve with representative volumes, e.g., with the minimum, 25th, 50th and 75th and maximum percentile. Moos et al.  $(2021)$  $(2021)$  sample eight RBSDs  $[m<sup>3</sup>]$  of different rockfall sites and fit them to a power law distribution of the form  $f(x, a, b) = ax^{-b}$ . Generally, the calculated block volumes of certain return periods are signifcantly larger than the expert-based maximum block volumes. The ftted power law only corresponds well to the empirical data for block volumes  $≥0.05$  m<sup>3</sup>. There is a strong influence of parameter *b* on the modelled frequency.

Today, Palmström's method of creating an IBSD by cutting a cube by joint sets can be applied in a more advanced way using synthetic rock mass (SRM) models. We asked whether IBSDs derived from SRM models can be described by probability distribution functions and if yes, how well. Because IBSDs are depending on the joint properties and they are often described by the lognormal and exponential distribution functions, we asked how well those functions describe IBSDs. Describing IBSDs by probability distribution functions could allow for more certain, accurate, verifable, holistic, and objective results. For this purpose, we calculate diferent synthetic rock mass (SRM) models based on photogrammetric surveys by UAV (unmanned aerial vehicle). We carry out exact rock mass block surveys with the help of these models and determine IBSDs. We ft the lognormal, exponential, and various other distribution functions to the determined IBSDs. Finally, we compare the correlations between the IBSDs and the distribution functions using the KS test and the MSE method.

### **2 Method**

We determine block size distributions from detachment areas (IBSDs) using synthetic rock mass (SRM) models, ft them to various distribution functions and check their correlations.

In SRM models, a discrete joint network (DFN) is intersected with a volume model to simulate the rock mass (Fig. [1\)](#page-3-0). The central element here is the DFN. It statistically describes the geometric properties of the joint sets. For this purpose, we consider the distribution of the joint orientation, the joint density (e.g., number of joints per m; reciprocal of joint spacing) and the joint size distribution (persistence) for each individual joint set. A DFN results in a collection of disk-shaped joints whose geometric properties, such as location, orientation, density/intensity, and joint size distribution, are subject to a probability distribution. Thus, the joints created in this way do not represent the actual joints in the rock mass. Nevertheless, in this way it is possible to model the joint system very realistically.

In detail, the procedure starts with a survey of the rock face by unmanned aerial vehicle (UAV). We took several high-resolution photos of the rock face from diferent angles and distances with respect to the rock face. Thanks to integrated GPS within the UAV, we know the positions of where the photographs are taken. To create a 3D model by photogrammetry, it is necessary to clearly assign each point of the rock face from at least three diferent perspectives. From the overlapping photos, the software Agisoft Metashape (Agisoft [2021](#page-24-0)) creates a point cloud and triangulates it into a mesh. This results in a 3D digital elevation model (DEM) of the rock face. We analysed the point cloud regarding the orientation and distance of the joint sets using the CloudCompare software (Cloudcompare [2020\)](#page-25-10). *Joint orientations* can be measured at the visible outcrops of the rock face. To measure orientations, CloudCompare creates a fat surface for a certain area if the correspondence of its containing points is large enough. The orientation of the created surface is specifed with dip angle and dip direction. We can group the measured joints into joint sets. It is not possible to measure the *joint spacings* in CloudCompare directly. These are defned as normal distances between two joints of the same joint set, which are generally neither parallel nor



<span id="page-3-0"></span>**Fig. 1** Intersection of a volume model with a discrete fracture network (DFN) creates a synthetic rock mass (SRM) model

at right angles to the rock face. However, by back-calculating to normal distances and averaging the measurements, we can derive joint spacings. *Joint size* (persistence) is another very important rock mass parameter. It can have a signifcant infuence on the strength and stability of the rock slope. Pahl ([1981\)](#page-25-11) has developed a method and equations to calculate the approximate average joint size of a joint set. His method requires a cut-of of small joint sizes and provides an exact solution, if the joint size distribution can be assumed to be exponential. There has been considerable discussion in the literature as to whether the distribution of joint sizes is a negative exponential or a lognormal distribution (Hudson and Harrison [1997\)](#page-25-7). However, in this case we are interested in block sizes of future rockfall events. They require full detachment from the rock face. Thus, we assume that the blocks are completely cut free by the intersecting joint sets. In other words, we assume 100% persistence for all joints.

Based on the above survey information, we can develop a discrete joint network (DFN). A DFN artifcially reproduces the existing joint structure as realistically as possible. The *joint orientations* are created by the bootstrapping method. This method assumes that the available random sample is 'representative' of the population from which it is drawn. The bootstrap replaces the theoretical distribution function of a random variable by the empirical distribution function of the sample. So, it is obvious that bootstrapping only works well if the empirical distribution function can approximate the actual distribution function sufficiently well. This requires a certain size of the original sample. Bootstrapping can be

understood as a Monte Carlo method, since it repeatedly draws random samples of a distribution. 3DEC ofers the possibility to generate the *joint spacings* indirectly via the joint density or joint intensity. The joint density is defned as the area of joints per unit volume  $[m<sup>2</sup>/m<sup>3</sup>]$  (P32), or the length of joints per unit area  $[m/m<sup>2</sup>]$  (P21), or the number of joints per unit length  $\lceil m^{-1} \rceil$  (P10). One square meter of joint in one cubic meter of the rock mass would correspond to a joint density of  $1 \, [\text{m}^2/\text{m}^3]$ . The joint density depends only on the joint area per volume. So, several small joint areas can have the same density as a few large joint areas within the same volume. 3DEC models *joint sizes* as disks. The joint length, or generally the joint size, refers to the diameter of this disk. We can set the limits of the smallest and largest disk  $(l_{\text{min}}$  and  $l_{\text{max}})$  in the software (Itasca [2020](#page-25-12)). In our case, we model disks that go through the entire model domain (100% persistence).

We simplifed the determination of the joint density by counting the number of joints per length normal to the joints (P10), for each joint set. Assuming a persistence of 100%, the number of joints per unit length corresponds to the area of joints per unit volume, i.e.,  $P10 = P21 = P32$ .

By intersecting the DEM with the DFN, we calculate the SRM model. We are not setting any strength parameters for the rock mass. We are only interested in the volume of the blocks. So, we can rather speak of a synthetic rock block model than a synthetic rock mass model. With the help of this model, we perform an exact rock mass block survey and determine a holistic IBSD.

The derived IBSDs represent block volumes  $[m<sup>3</sup>]$ . We want to check their correlations to various ftted continuous distributions by the Kolmogorov–Smirnov (KS) test and the mean-squared error (MSE) method. We are not interested in properties like density or porosity, which would be afected by cubic dimensions. In our case, it seems more appropriate to work with size distributions in linear dimensions [m] rather than in cubic dimensions  $[m<sup>3</sup>]$ . Furthermore, fitting distribution functions to linear dimensions can provide a more intuitive visual representation of the size distribution (e.g., grain sizes in a sieve curve). For these reasons, we are transferring the derived IBSDs from cubic meters to meters by taking the cube root.

We use Python to check our derived IBSDs against  $79<sup>1</sup>$  $79<sup>1</sup>$  $79<sup>1</sup>$  distribution functions (Christopher [2017](#page-24-1)). Among them are also the lognormal and exponential distribution functions.

The probability density function (pdf) for lognorm is:

$$
f(x,s) = \frac{1}{sx\sqrt{2\pi}}exp\left(-\frac{\log^2(x)}{2s^2}\right)
$$
 (1)

for  $x > 0$ ,  $s > 0$ , where x is the random variable and s is the standard deviation. Lognorm takes *s* as a shape parameter. The probability density above is defned in the 'standardized' form. To shift and/or scale the distribution, the loc and scale parameters are used. Specifcally, lognorm.pdf(*x*,*s*, loc, scale) is identically equivalent to  $\frac{\text{lognorm.pdf}(y,s)}{\text{scale}}$  with  $y = \frac{x-\text{loc}}{\text{scale}}$ . In

<span id="page-4-0"></span><sup>&</sup>lt;sup>1</sup> Norm, alpha, anglit, arcsine, beta, betaprime, bradford, burr, cauchy, chi, chi2, cosine, dgamma, dweibull, erlang, expon, exponweib, exponpow, fatiguelife, foldcauchy, f, fsk, foldnorm, gamma, gausshyper, genexpon, genextreme, gengamma, genlogistic, genpareto, genhalfogistic, gilbrat, gompertz, gumbel\_l, gumbel\_r, halfcauchy, halfogistic, halfnorm, hypsecant, invgamma, invweibull, johnsonsb, johnsonsu, laplace, logistic, loggamma, loglaplace, lognorm, lomax, maxwell, mielke, nakagami, ncx2, ncf, nct, norm, pareto, powerlaw, powerlognorm, powernorm, rdist, reciprocal, rayleigh, rice, recipinvgauss, semicircular, t, triang, truncexpon, truncnorm, tukeylambda, uniform, vonmises, wald, weibull\_min, weibull\_max, wrapcauchy, ksone, kstwobign.

case of a normally distributed random variable *x* with mean *m* and standard deviation *s*,  $y = exp(x)$  is normally distributed with standard deviation *s* and scale =  $exp(m)$ .

The probability density function (pdf) for expon is:

$$
f(x) = \exp(-x) \tag{2}
$$

for  $x \ge 0$ . The probability density above is defined in the 'standardized' form. To shift and/or scale the distribution, the loc and scale parameters are used. Specifcally, expon.pdf(*x*, loc, scale) is identically equivalent to  $\frac{\text{expon.pdf}(y)}{\text{scale}}$  with  $y = \frac{x - \text{loc}}{\text{scale}}$ . A common parametrization for expon is in terms of the rate parameter  $\lambda$ , such that  $pdf = \lambda \cdot exp(-\lambda \cdot x)$ . This parametrization corresponds to using scale  $=$   $\frac{1}{\lambda}$ .

The Kolmogorov–Smirnov test, also known  $\hat{a}$  the KS test, is a nonparametric statistical test. It assesses whether a sample comes from a specifc distribution. In other words, the test compares the observed data with the predicted data (also known as the one-sample KS test) under the null hypothesis that the two distributions are identical. It is sensitive to both location and shape diferences between the sample and reference distributions. The KS test results in the test statistic  $(D)$  and the *p* value. These two components are essential in interpreting the outcome of the KS test. The test statistic (*D*) is a numerical value that compares the sum of vertical distances of all data points (supremum) of the cumulative distribution function (cdf) of the sample to the sum of vertical distances of all data points of the cdf of the reference distribution. The *D* value ranges between 0 and 1. A value of 0 means a perfect match between the sample and reference distribution. The  $p$  value is a probability that measures the strength of evidence against the null hypothesis. It determines whether the observed diference (*D*) is meaningful and reliable or can be attributed to random chance. If the *p* value is smaller than the chosen significance level (e.g.,  $p \le 0.05$ ), we can consider the result statistically signifcant. We can reject the null hypothesis in favour of concluding that the sample does not follow the reference distribution. If the *p* value is greater than the signifcance level, the result is not statistically signifcant. We cannot reject the null hypothesis. There is no sufficient evidence to claim a difference between the sample and the reference distribution. We choose a confdence level of 95%; that is, we reject the null hypothesis in favour of the alternative if the  $p$  value is less than 0.05 (i.e.,  $5\%$ ).

The mean-squared error (MSE) is a statistical metric, also known as the meansquared deviation. It measures the average squared diferences between the observed values and the predicted values in a dataset. It quantifes the accuracy of a prediction by assessing how well it fts the observed data points. For each data point, the squared diference between the observed value and the predicted value is calculated. Squaring the diferences ensures that negative diferences do not cancel out positive diferences. All squared diferences are summed up and divided by the total number of data points. This results in the average squared diference, which is the MSE. Lower MSE values indicate a better ft of the model to the data (Hedderich and Sachs [2020\)](#page-25-13).

#### **3 Results**

We applied the described method to three diferent rock faces in three diferent areas within Lower Austria: Tiefenbach, Spitz and Greifenstein, as shown in Fig. [2](#page-6-0).



<span id="page-6-0"></span>**Fig. 2** Project areas (from West to East) Tiefenbach, Spitz and Greifenstein (black) within Lower Austria (Gba [2002\)](#page-25-15)



<span id="page-6-1"></span>**Fig. 3** Geological map (1:50 000) of the project area Tiefenbach. The red star marks the location of the rock face. The geological unit is the so-called Weinsberger Granite (38): coarse-grained biotite granite with porphyritic large potassium feldspar (Mississipium). The pink Xs mark fne-grained granitic dikes (Moser and Linner [2019\)](#page-25-14)

#### **3.1 Tiefenbach**

The rock face in Tiefenbach is in the West of Lower Austria, along the state road Greiner Straße (B 119) next to and along the Danube, which represents the border to Upper Austria there. The rock face is about 25 m high. The geological unit (Fig. [3\)](#page-6-1) is the so-called Weinsberger Granite (38): coarse-grained biotite granite with porphyritic large potassium feldspar (Mississipium) next to metablastic to dialectic paragneiss (53), relics of biotiterich paragneiss ('pearl gneiss') (Moser and Linner [2019\)](#page-25-14).

The 3D photogrammetry model of the Tiefenbach rock face is shown in Fig. [4l](#page-7-0)eft. We limited the structural geological analysis of the Tiefenbach rock face to the marked area



<span id="page-7-0"></span>**Fig. 4** Left: 3D photogrammetry model (Agisoft [2021](#page-24-0)) of the Tiefenbach rock face. The structural geological analysis was limited to the marked area (red); Right: 3D point cloud model (Cloudcompare [2020](#page-25-10)) of the selected Tiefenbach rock face, coloured according to the dip direction (degrees): *k*1 (red), *k*2 (blue/yellow), *k*3 (green)



<span id="page-7-1"></span>**Fig. 5** Equal-area lower hemisphere plot of the 102 measured joints (great circles) and the slope (black) in Tiefenbach, using OpenStereo (Grohmann and Companha [2017](#page-25-17)); for the density plot, refer to Fig. [7](#page-8-2)(left)

in Red, because of the low vegetation outcrop. The investigated rock face has a length of about 38 m, a slope height of 20–25 m and an average inclination of 73°. Figure [4](#page-7-0)right shows the 3D point cloud model of the selected Tiefenbach rock face coloured according to the dip direction.

The analysis via CloudCompare resulted in 102 joint orientations (Helm [2023](#page-25-16)). We grouped them into three joint sets: *k*1 (red), *k*2 (blue and yellow) and *k*3 (green) with 30, 58 and 14 measured orientations, respectively (Fig. [5](#page-7-1)). We used all measured joint orientation to generate the DFN. They provide a sufficiently large sample for bootstrapping. Table [1](#page-8-0) lists the mean dip direction and dip, and the determined joint densities for each joint set.

The dimensions  $(x \ y \ z)$  of the model domain are approx. 230×500×100 m. We cut the model domain by two planes parallel to the slope (289/73) with a 50-m distance. The resulting 3D volume model of the slope is approx. 500 m long, 50 m deep (into the rock mass) and 100 m high. Figure [6](#page-8-1) shows the joint sets of the rock mass slope in Tiefenbach, with the *y* axis pointing north. Figure [7](#page-8-2) compares the equal-area lower hemisphere density

<span id="page-8-0"></span>**Table 1** Tiefenbach DFN data: joint directions and joint density





<span id="page-8-1"></span>**Fig. 6** 3DEC SRM model of Tiefenbach showing the block joint sets (*y* axis=north)



<span id="page-8-2"></span>**Fig. 7** Equal-area lower hemisphere density plots of the 102 measured joint poles (left; using OpenStereo; max~337/76 and~069/87) and the 407 created joint poles in the SRM model (right; using 3DEC; max.~162/77) for Tiefenbach

plots of the 102 measured joint poles [left; using OpenStereo (Grohmann and Companha [2017\)](#page-25-17)] and the 94 created joint poles in the SRM model (right) for Tiefenbach.

We derived 60,503 block volumes from the Tiefenbach SRM model. Assuming the block shape cuboid, we calculated their edge lengths by taking the cube root. The IBSD



<span id="page-9-0"></span>**Fig. 8** IBSD of the Tiefenbach slope with 60,503 blocks (edge length *a* in [m])



<span id="page-9-1"></span>**Fig. 9** IBSD of the Tiefenbach slope [m] (blue bars) with the ftted exponential (maroon), lognormal (blue) and generalized exponential (green) probability density functions (left: pdfs, right: cdfs)



[m] of the Tiefenbach slope is plotted in Fig. [8.](#page-9-0) Our check against 79 diferent distribution functions with Python found that neither the exponential nor the lognormal distribution fts well. Rather, the generalized exponential distribution function fts best. We show this both graphically in Fig. [9](#page-9-1) and numerically using the KS test and the mean-squared error (MSE)

method (as described above).

<span id="page-9-2"></span>**Table 2** Fitting parameters of the ftted distribution functions and computational test results of the KS tests and MSE method; SRM

Tiefenbach

The ftted parameters of the three tested continuous distribution functions and the KS test and MSE results are listed in Table [2](#page-9-2). With a *p* value near 0.00, we reject the null hypothesis in favour of concluding that the sample does not follow the reference distributions. This is the case for the exponential (expon) and the lognormal (lognorm) distribution.

Ouantile	DFN $[m^3]$	$DFN$ [m]	Expon $[m]$	Lognorm [m]	Genexpon [m]
$\mathbf{0}$	$3.46e - 12$	$1.51e - 04$	$1.51e - 04$	$-3.19e - 01$	$1.51e - 04$
25	$2.95e - 01$	0.67	0.53	0.70	0.67
50	3.19	1.47	1.29	1.36	1.46
75	18.13	2.63	2.58	2.46	2.63
95	119.53	4.93	5.57	5.40	4.94
96	144.90	5.25	5.98	5.87	5.24
97	179.02	5.64	6.52	6.50	5.62
98	228.30	6.11	7.27	7.44	6.15
99	340.97	6.99	8.56	9.18	7.04
100	5973.04	18.14	Inf	Inf	Inf

<span id="page-10-0"></span>**Table 3** Comparison of quantiles for the SRM Tiefenbach

The percentiles printed in bold are those selected in accordance with the ONR 24810 guideline, depending on the event frequency



<span id="page-10-1"></span>**Fig. 10** Geological map (1:50 000) of the project area Spitz. The red star marks the location of the rock face. The geological unit is marble, ribbon marble and silicate marble (46) (Fuchs et al. [1983\)](#page-25-18)

With a  $p$  value of 0.58 ( $> 0.05$ ), we cannot reject the null hypothesis. This indicates that there is no sufficient evidence to claim a difference between the sample and the generalized exponential (genexpon) reference distribution. Comparing the MSE values for the expon, lognorm and genexpon distribution functions, the genexpon distribution function has the lowest MSE value, indicating its best ft. A comparison of the quantiles is listed in Table [3](#page-10-0).

#### **3.2 Spitz**

The rock face in Spitz (Fig. [10\)](#page-10-1) is a former marble quarry in the middle of Lower Austria, West of the state road Donau Straße (B 3) next to and along the Danube. The geo-logical unit (Fig. [10](#page-10-1)) is marble, ribbon marble and silicate marble (46) (Fuchs et al. [1983](#page-25-18)). The 3D photogrammetry model of the Spitz rock face is shown in Fig. [11t](#page-11-0)op.



<span id="page-11-0"></span>**Fig. 11** Top: 3D photogrammetry model (Agisoft [2021\)](#page-24-0) of the Spitz rock face; bottom: 3D point cloud model of the selected Spitz rock face, coloured according to the dip direction (degrees): SE (green), N/S (red/cyan), NW (purple)

The investigated rock face has a length of about 70 m, a slope height of up to 30 m and an average inclination of 84°. Figure [11b](#page-11-0)ottom shows the 3D point cloud model of the selected Spitz rock face coloured according to the dip direction.

The analysis via CloudCompare resulted in three main joint sets forming the rock face: SE, N and NW with 132, 55 and 21 (total 208) measured orientations, respectively (Fig. [12\)](#page-12-0). To generate the DFN, all measured joint orientations are used to provide a sufficient large sample for bootstrapping. Table [4](#page-12-1) lists the mean dip direction and dip, as well as the determined mass densities and joint size for each joint set.

The dimensions  $(x \ y \ z)$  of the model domain are approx.  $500 \times 100 \times 100$  m. We cut the model domain by two planes parallel to the slope (356/84) with a 50-m distance. The resulting 3D volume model of the slope is approx. 500 m long, 50 m deep (into the rock mass) and 100 m high. Figure [13](#page-12-2) shows the joint sets of the rock mass slope in Spitz, with the *y* axis pointing north. Figure [14](#page-13-0) compares the equal-area lower hemisphere



<span id="page-12-0"></span>**Fig. 12** Equal-area lower hemisphere plot of the 208 measured joints (great circles) and the slope (black) in Spitz, using OpenStereo; for the density plot, refer to Fig. [14](#page-13-0) (left)

<span id="page-12-1"></span>



<span id="page-12-2"></span>**Fig. 13** 3DEC SRM model of Spitz showing the block joint sets (*y* axis=north)

density plots of the 208 measured joint poles (left) and the 174 created joint poles in the SRM model (right) for Spitz.

We derived 26,253 block volumes from the Spitz SRM model. Assuming the block shape cuboid, we calculated their edge lengths by taking the cube root. The IBSD [m] of the Spitz slope is plotted in Fig. [15](#page-13-1). Our check against 79 diferent distribution functions



<span id="page-13-0"></span>**Fig. 14** Equal-area lower hemisphere density plots of the 208 measured joint poles (left; using OpenStereo; max~356/85 and~135/40) and the 185 created joint poles in the SRM model (right; using 3DEC; max.~303/80) for Spitz



<span id="page-13-1"></span>**Fig. 15** IBSD of the Spitz slope with 26,253 blocks (edge length *a* in [m])



<span id="page-13-2"></span>**Fig. 16** IBSD of the Spitz slope [m] (blue bars) with the ftted exponential (maroon), lognormal (blue) and generalized exponential (green) probability density functions (left: pdfs, right: cdfs)

with Python found that neither the exponential nor the lognormal distribution fts well. Rather, the generalized exponential distribution function fts best. We show this both graphically in Fig. [16](#page-13-2) and numerically using the KS test and the mean-squared error (MSE) method (as described above).

<span id="page-14-0"></span>

<b>Ouantile</b>	DFN $[m^3]$	$DFN$ [m]	Expon $[m]$	Lognorm [m]	Genexpon [m]
$\mathbf{0}$	$8.45e-13$	$9.45e - 05$	$9.45e - 05$	$-4.07e - 01$	$9.45e - 05$
25	0.70	0.89	0.70	0.93	0.88
50	6.94	1.91	1.70	1.80	1.94
75	41.21	3.45	3.40	3.24	3.48
95	275.69	6.51	7.34	7.11	6.49
96	332.99	6.93	7.88	7.73	6.87
97	411.60	7.44	8.59	8.55	7.35
98	544.03	8.16	9.58	9.79	8.02
99	792.58	9.25	11.28	12.09	9.13
100	3776.36	15.57	Inf	Inf	Inf

<span id="page-14-1"></span>**Table 6** Comparison of quantiles for the SRM Spitz

The percentiles printed in bold are those selected in accordance with the ONR 24810 guideline, depending on the event frequency

The ftted parameters of the three tested continuous distribution functions and the KS test and MSE results are listed in Table  $5$ . With a  $p$  value near 0.00, we reject the null hypothesis in favour of concluding that the sample does not follow the reference distributions. This is the case for the exponential (expon) and the lognormal (lognorm) distribution. With a *p* value of 0.16 ( $> 0.05$ ), we cannot reject the null hypothesis. This indicates that there is no sufficient evidence to claim a difference between the sample and the generalized exponential (genexpon) reference distribution. Comparing the MSE values for the expon, lognorm and genexpon distribution functions, the genexpon distribution function has the lowest MSE value, indicating its best ft. A comparison of the quantiles is listed in Table [6](#page-14-1).

#### **3.3 Greifenstein**

The rock face in Greifenstein is within a former quarry (until 1993) and landfll North of Vienna, also close to the Danube, located within the Rhenodanubian fysch zone (Figs. [17,](#page-15-0) [18](#page-15-1)). This zone extends from Vienna to Vorarlberg north of the northern Limestone Alps. Basically, the fysch zone was formed by sedimentary depositional processes of rivers in the sea of that time. Due to the diferent fow velocities, sedimentary



<span id="page-15-0"></span>**Fig. 17** Top: 3D photogrammetry model (Agisoft [2021](#page-24-0)) of the Greifenstein rock face; Bottom: 3D point cloud model of the selected Greifenstein rock face, coloured according to the dip direction (degrees): bedding (light blue), N (red), SW/NE (blue/yellow)



<span id="page-15-1"></span>**Fig. 18** Geological map (1:50 000) of the project area Greifenstein. The red star marks the location of the rock face. The geological unit is the so-called Greifenstein Formation (58): fne- to coarse-grained, medium- to thick-bedded, siliciclastic sandstone and clay shale (Ypresium) (Kreuss [2020\)](#page-25-20)

layers of diferent thickness were formed. About 42 million years ago, the 'Eurasian Plate' and the 'Adriatic Plate' collided and the fysch zone came to the surface (Egger and Coric [2017](#page-25-19)). The easternmost part of this Rhenodanubian fysch zone is called the



<span id="page-16-0"></span>**Fig. 19** Equal-area lower hemisphere plot of the measured joints (great circles) and the slope (black) in Greifenstein, using OpenStereo; for the density plot refer to Fig. [21](#page-17-1) (left)

<span id="page-16-1"></span>**Table 7** Greifenstein DFN data: joint directions and joint density



Greifenstein Formation (58): fne- to coarse-grained, medium- to thick-bedded, siliciclastic sandstone and clay shale (Ypresium) (Kreuss [2020\)](#page-25-20).

The 3D photogrammetry model of the Greifenstein rock face is shown in Fig. [17](#page-15-0)top. The investigated rock face has a length of about 80 m, a slope height of 20–27 m and an average inclination of 70° (Wiesinger [2023](#page-25-21)). Figure [17b](#page-15-0)ottom shows the 3D point cloud model of the selected Greifenstein rock face coloured according to the dip direction.

The structural geological analysis of the Greifenstein 3D rock face model resulted in 143 joint orientations, which could be grouped into three joint sets: the bedding, the Northeast-Southwest joint set and the North joint set (Fig. [19\)](#page-16-0) (Wiesinger [2023\)](#page-25-21). To generate the DFN, all measured joint orientations are used to provide a sufficient large sample for bootstrapping. Table [7](#page-16-1) lists the mean dip direction and dip, as well as the determined mass densities and joint size for each joint set.

The dimensions  $(x \, y \, z)$  of the model domain are approx.  $200 \times 160 \times 40$  m. We cut the model domain by two planes parallel to the slope (011/70) with a 20-m distance. The resulting 3D volume model of the slope is approx. 200 m long, 20 m deep (into the rock mass) and 40 m high. Figure [20](#page-17-0) shows the joint sets of the rock mass slope in Greifenstein, with the *y* axis pointing north. Figure [21](#page-17-1) compares the equal-area lower hemisphere density plots of the 143 measured joint poles (left) and the 381 created joint poles in the SRM model (right) for Greifenstein.

We derived 143,029 block volumes from the Greifenstein SRM model. Assuming the block shape cuboid, we calculated their edge lengths by taking the cube root. The IBSD [m] of the Greifenstein slope is plotted in Fig. [22.](#page-17-2) Our check against 79 diferent distribution



<span id="page-17-0"></span>**Fig. 20** 3DEC SRM model of Greifenstein showing the block joint sets (*y* axis=north)



<span id="page-17-1"></span>**Fig. 21** Equal-area lower hemisphere density plots for Greifenstein; of the 143 measured joint poles (left; using OpenStereo; max.~357/65); and the 381 created joint poles in the SRM model (right; using 3DEC;  $max. -135/23$ 



<span id="page-17-2"></span>**Fig. 22** IBSD of the Greifenstein slope with 143,029 blocks (edge length *a* in [m])



<span id="page-18-0"></span>**Fig. 23** IBSD of the Greifenstein slope [m] (blue bars) with the ftted exponential (maroon), lognormal (blue) and generalized exponential (green) probability density functions (left: pdfs, right: cdfs)



computational test results of the KS tests and MSE method; SRM Greifenstein

<span id="page-18-1"></span>**Table 8** Fitting parameters of the ftted distribution functions and

functions with Python found that neither the exponential nor the lognormal distribution fts well. Rather, the generalized exponential distribution function fts best. We show this both graphically in Fig. [23](#page-18-0) and numerically using the KS test and the mean-squared error (MSE) method (as described above).

The ftted parameters of the three tested continuous distribution functions and the KS test and MSE results are listed in Table [8.](#page-18-1) With a *p* value near 0.00, we reject the null hypothesis in favour of concluding that the sample does not follow the reference distributions. This is the case for the exponential (expon) and the lognormal (lognorm) distribution. With a *p* value of  $0.58$  ( $> 0.71$ ), we cannot reject the null hypothesis. This indicates that there is no sufficient evidence to claim a diference between the sample and the generalized exponential (genexpon) reference distribution. Comparing the MSE values for the expon, lognorm and genexpon distribution functions, the genexpon distribution function has the lowest MSE value, indicating its best ft. A comparison of the quantiles is listed in Table [9](#page-19-0).

Additionally, we have examined whether there is a diference between the block size distribution derived from a rock slope and a block size distribution derived from the rock mass (i.e., a cube) in the example of Greifenstein. The dimensions  $(x \ y \ z)$  of the model domain are approx.  $60 \times 60 \times 60$  m. Figure [24](#page-19-1) shows the joint sets of the rock mass cube in Greifenstein, with the y-axis pointing north. 134,711 block volumes were derived from the SRM model, and assuming the block shape cuboid, their edge lengths were calculated taking the cube root. The distributions of the block sizes are plotted in Fig. [25](#page-19-2). Again, neither the exponential nor lognormal distributions ft well. Rather, the generalized exponential

Ouantile	DFN $\lceil m^3 \rceil$	$DFN$ [m]	Expon $[m]$	Lognorm [m]	Genexpon [m]	
$\mathbf{0}$	$2.42e-15$	$1.34e - 0.5$	$1.34e - 0.5$	$-0.07$	$1.34e - 0.5$	
25	$7.45e - 03$	0.20	0.16	0.20	0.19	
50	$7.45e - 02$	0.42	0.39	0.40	0.42	
75	0.46	0.77	0.78	0.73	0.77	
95	3.87	1.57	1.68	1.69	1.57	
96	4.77	1.68	1.81	1.85	1.68	
97	6.14	1.83	1.97	2.06	1.82	
98	8.36	2.03	2.20	2.38	2.02	
99	13.11	2.36	2.59	2.99	2.36	
100	225.20	6.08	Inf	Inf	Inf	

<span id="page-19-0"></span>**Table 9** Comparison of quantiles for the SRM Greifenstein

The percentiles printed in bold are those selected in accordance with the ONR 24810 guideline, depending on the event frequency



<span id="page-19-1"></span>**Fig. 24** 3DEC SRM model of the Greifenstein 60 m cube showing the block joint sets (*y* axis = north)



<span id="page-19-2"></span>**Fig. 25** IBSD of Greifenstein 60 m cube with 134,711 blocks (edge length *a* in [m])

distribution function fts best. We show this both graphically in Fig. [26](#page-20-0) and numerically using the KS test and the mean-squared error (MSE) method (as described above).

The ftted parameters of the three tested continuous distribution functions and the KS test and MSE results are listed in Table [10.](#page-20-1) With a *p* value near 0.00, we reject the null hypothesis

<span id="page-20-1"></span>**Table 10** Fitting parameters of the ftted distribution functions and computational test results of the KS tests and MSE method; SRM Greifenstein 60 m cube



<span id="page-20-0"></span>**Fig. 26** IBSD of the Greifenstein 60 m cube [m] (blue bars) with the ftted exponential (maroon), lognormal (blue) and generalized exponential (green) probability density functions (left: pdfs, right: cdfs)

Fitting parameters	Expon	Lognorm	Genexpon
Loc	$8.55e - 0.5$	$-1.34e-01$	$8.55e - 0.5$
scale	0.71	0.66	0.96
shape par. 1		$s = 0.71$	$a = 0.92$
shape par. 2			$b = 1.48$
shape par. 3			$c = 0.66$
KS: p	0.00	$4.03e - 133$	0.09
KS: D	$6.42e - 02$	$3.37e - 02$	$3.38e - 03$
<b>MSE</b>	3.82	6.15	1.48

<span id="page-20-2"></span>**Table 11** Comparison of quantiles for the SRM Greifenstein cube 60 m



The percentiles printed in bold are those selected in accordance with the ONR 24810 guideline, depending on the event frequency

in favour of concluding that the sample does not follow the reference distributions. This is the case for the exponential (expon) and the lognormal (lognorm) distribution. With a *p* value of  $0.09$  ( $> 0.05$ ) we cannot reject the null hypothesis. This indicates that there is no sufficient evidence to claim a diference between the sample and the generalized exponential (genexpon)

reference distribution. Comparing the MSE values for the expon, lognorm and genexpon distribution functions, the genexpon distribution function has the lowest MSE value, indicating its best ft. A comparison of the quantiles is listed in Table [11.](#page-20-2)

#### **4 Discussion**

The equal-area density plots proof that it is possible to represent joint systems using a DFN. The joint orientations of the left and right density plots correspond very well (see Figs. [7,](#page-8-2) [14](#page-13-0), [21](#page-17-1)). The pole densities of the measured joint planes (left plots) represent joints exposed on the rock face. So, wall-building joints may be measured more frequently than other joints. This can lead to a situation where joint surfaces of lower joint density (with greater joint distances) are measured more frequently on the rock face, and vice versa. Thus, measurements from the rock face may be distorted. The SRM density plots (on the right) refect the 'true' joint densities of the joint sets (relative to each other). For example, in Greifenstein (Fig. [21](#page-17-1), left) the joint set N (357/65, red) is dominating on the rock face and measured more frequently than the bedding (135/23, light blue). The surface areas (outcrops) of the bedding are very small. Yet, the bedding has a much higher joint density than the other two joint sets. This is refected in the density plot in Fig. [21](#page-17-1) on the right. We can conclude that the density plot of the SRM is more realistic and more meaningful than the density plot of the measured joints on the rock face.

We chose the slope and cube dimensions of the SRM models to create many blocks for our investigations. In Tiefenbach, a slope size of approximately  $500 \times 50 \times 100$  m generated 60,503 block volumes. In Spitz, a similar slope size generated 26,253 block volumes. In Greifenstein, with a bedding of relatively high joint density, a slope size of approximately  $200\times20\times40$  m generated 143,029 block volumes, and the Greifenstein 60 m cube generated 134,711 block volumes. The size of the model should correspond to the homogeneous area investigated.

We show that neither the lognormal nor the exponential distribution functions describe IBSDs [m] well. To ft best, the lognormal is shifted with a negative loc parameter. This results in negative block volumes. Both, the lognorm and expon have relatively long tails. This results in much larger blocks compared to the IBSDs. We also tried to ft Python's power law distribution in the form  $f(x, a) = ax^{a-1}$  to our derived IBSDs (both in [m<sup>3</sup>] and in [m]). No correlations could be found. The generalized exponential distribution function best describes block size distributions [m] across three various lithologies when compared to 78 other distribution functions via the one-sample KS test and the MSE method.

The probability density function (pdf) for genexpon is:

$$
f(x, a, b, c) = (a + b(1 - \exp(-cx))) \exp\left(-ax - bx + \frac{b}{c}(1 - \exp(-cx))\right)
$$
(3)

for  $x \ge 0$ ,  $a, b, c > 0$ , where x is the random variable and a, b, and c are shape parameters. The probability density above is defned in the 'standardized' form. To shift and/or scale the distribution, the loc and scale parameters are used. Specifcally, genexpon.pdf(*x*, *a*, *b*, *c*, loc, scale) is identically equivalent to  $\frac{\text{generagon\_pdf}(y, a, b, c)}{\text{scale}}$  with  $y = \frac{x - \text{loc}}{\text{scale}}$ . Above generalized exponential distribution is an extension of Marshall and Olkin's bivariate exponential distribution (Ryu [1993\)](#page-25-22). The three shape parameters provide quite a bit of fexibility for analysing any skewed dataset. Table [12](#page-22-0) lists the genexpon ftting parameters

Genexpon fitting	<b>SRM</b> Tiefenbach	<b>SRM Spitz</b>	<b>SRM</b> Greifenstein		
parameters	$500 \times 50 \times 100$	$500 \times 50 \times 100$	$200 \times 20 \times 40$	Cube $60 \text{ m}$	
loc	$1.51e - 04$	$9.45e - 05$	$1.34e - 0.5$	$8.55e - 0.5$	
scale	2.48	2.20	0.71	0.96	
shape par. 1	$a = 0.96$	$a = 0.66$	$a = 0.87$	$a = 0.92$	
shape par. 2	$b = 1.14$	$b = 1.01$	$b = 0.58$	$b = 1.48$	
shape par. 3	$c = 0.73$	$c = 0.32$	$c = 2.97$	$c = 0.66$	

<span id="page-22-0"></span>**Table 12** Genexpon fitting parameters for the investigated locations Tiefenbach (slope size  $500 \times 50 \times 100$ ) m), Spitz (slope size  $500 \times 50 \times 100$  m) and Greifenstein (slope size  $200 \times 20 \times 40$  m and 60 m cube)

<span id="page-22-1"></span>**Table 13** Comparison of quantiles [m] for diferent SRM models

Ouantile	Tiefenbach slope		Spitz slope		Greifenstein slope		Greifenstein cube	
	<b>DFN</b>	Genexpon	<b>DFN</b>	Genexpon	<b>DFN</b>	Genexpon	<b>DFN</b>	Genexpon
$\Omega$	$1.51e - 04$	$1.51e - 04$	$9.45e - 0.5$	$9.45e - 0.5$	$1.34e - 05$	$1.34e - 05$	$8.55e - 0.5$	$8.55e - 0.5$
25	0.67	0.67	0.89	0.88	0.20	0.19	0.26	0.26
50	1.47	1.46	1.91	1.94	0.42	0.42	0.56	0.57
75	2.63	2.63	3.45	3.48	0.77	0.77	1.00	1.00
95	4.93	4.94	6.51	6.49	1.57	1.57	1.85	1.84
96	5.25	5.24	6.93	6.87	1.68	1.68	1.96	1.95
97	5.64	5.62	7.44	7.35	1.83	1.82	2.10	2.08
98	6.11	6.15	8.16	8.02	2.03	2.02	2.29	2.27
99	6.99	7.04	9.25	9.13	2.36	2.36	2.61	2.58
100	18.14	Inf	15.57	Inf	6.08	Inf	5.67	Inf

The percentiles printed in bold are those selected in accordance with the ONR 24810 guideline, depending on the event frequency

for the diferent investigated slope and cube sizes. The location parameters are essentially zero.

For the Greifenstein slope and the Greifenstein cube calculations, we used the same DFN data (Table [7\)](#page-16-1). The models difer in their dimensions and orientation. We cut the slope model parallel to the mean rock face orientation. The cube model is north–south oriented, independent of the observed rock face orientation. Comparing the results (Table [12](#page-22-0)), the genexpon fitting parameters, especially  $b$  and  $c$ , do vary. It is unclear how these differences afect rockfall modelling results. Modelling a cube, independent of the slope direction, would be much more practicable when investigating a quarry, for example. Further investigations on the sensitivity of the scale and shape parameters are required. Comparing the quantiles of the Greifenstein slope and the Greifenstein cube models (Table [13\)](#page-22-1), the blocks of the cube model are generally slightly bigger. Due to the bigger dimensions of the cube in the *y* direction (20 m vs. 60 m), bigger blocks are built. The maximum block volumes depend on the size of the homogeneous area, as also recognized by Laimer [\(2019](#page-25-3)) and Moos et al. ([2021\)](#page-25-9).

The DFN- and genexpon-quantiles correspond very well (see Table [13\)](#page-22-1). Comparing the P98 of the genexpon cdfs, the block sizes seem plausible, considering the diferent



**Fig. 27** Photographs of the deposition areas; left: Spitz with huge marble blocks versus right: Greifenstein with smaller sandstone blocks due to rather dense bedding

<span id="page-23-0"></span>lithologies and joint systems. In Spitz, where we could observe huge marble blocks in the deposition area (Fig. [27](#page-23-0)left), the P98 block edge length in the genexpon cdf is 8 m  $(512 \text{ m}^3)$ . In Greifenstein, where we could observe a sandstone bedding of relatively high joint density (Fig. [27r](#page-23-0)ight), the P98 block edge length in the genexpon cdf is approximately 2 m  $(8 \text{ m}^3)$ . The P98 block edge length in the genexpon cdf of the Tiefenbach granite lies in between, with approximately 6 m  $(216 \text{ m}^3)$ .

As already criticized by Laimer ([2019\)](#page-25-3), the use of P95–P98 appears too high for rock formations, which form very large rockfall blocks  $(>10 \text{ m}^3)$ . He had sufficient data from the Dachstein Formation (limestones and dolomites) to show that the return periods of P95 to P96 blocks  $(0.15-2.25 \text{ m}^3)$  range from 23.5 to 56.5 years for this formation. This corresponds to the service life of a conventional rockfall protection barrier. P97 and P98 limestone blocks have return periods of more than 100 years.

In our holistic rock mass block investigations of relatively large rock masses (homogeneous areas), the problem of very large (P95–P98) blocks becomes even more apparent. Our SRM approach does not consider whether blocks are kinematically capable of failure. IBSDs can include blocks with very high return periods. One approach to deal with this problem could be a kinematic analysis of the blocks using block theory (Goodman and Shi [1985\)](#page-25-23). However, this approach does not consider return periods. In the design of rockfall protection measures and hazard analyses, rockfall frequencies (magnitude to frequency relations M/F (Corominas et al. [2018\)](#page-25-24)) and return periods play an important role. This requires knowledge of the events on the one hand and the defnition of a worst-case scenario (depending on the protection target) on the other. Our IBSD provides knowledge of all possible events within a homogenous area, including their frequencies (i.e., magnitude to frequency relations). Based on a defned worst-case scenario, events/block sizes of higher return periods may be neglected (cut of). This could be achieved based on (limited) available information, such as silent witnesses with estimated ages, reports from residents or records of past events.

We were able to find a probability distribution function that describes in situ block size distributions (IBSDs) very well. It could be generally applicable for block size distributions (BSDs). Further investigations on the sensitivity of the scale and shape parameters are required. We show that both the lognormal and the exponential distribution functions do not describe IBSDs well enough. Describing IBSDs by probability distribution functions could provide more certain, accurate, verifable, holistic, and objective results. With the presented method, it is possible to determine IBSDs based on photogrammetry and SRM models. The investigation of many more sites of diferent lithologies with this method could lead to a catalogue that recommends a range of scale and shape parameters for specific lithologies in the future. This requires locations of sufficiently large outcrops with low vegetation. The implications on rockfall modelling should be further investigated, for example, by comparing runout, kinetic energies, and bounce heights when modelling whole IBSDs versus genexpon pdfs and design blocks. The use of a distribution function together with a catalogue of suitable ftting parameters can ofer the advantage that the determination of an IBSD, and thus, a meaningful evaluation of a design block is also possible with a limited number of block size measurements (silent witnesses).

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**Author contributions** MI and AP did conceptualization, writing—review and editing and methodology; MI done formal analysis and investigation and writing—original draft preparation; AP supervised the study.

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# **Declarations**

**Confict of interest** The authors have no relevant fnancial or non-fnancial interests to disclose.

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