

INTEGRATING LASER SCANNING WITH DISCRETE ELEMENT MODELING FOR IMPROVING SAFETY IN UNDERGROUND STONE MINES

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Determination of Stopping Methodology for Mining Secondary Stopes by FLAC3D



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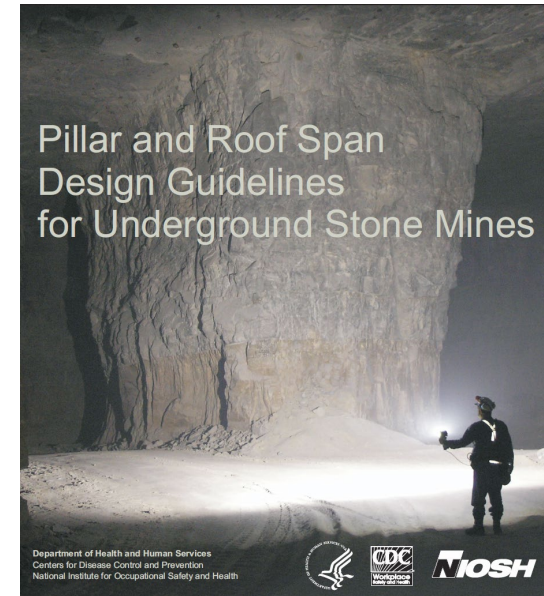
I. INTRODUCTION

“The guidelines for pillar and roof span design are empirically based; their validity, therefore, is restricted to rock conditions, mining dimensions, and pillar stresses that are similar to those included in this study. These guidelines should be applicable to the majority of stone mines in the Eastern and Midwestern United States. If pillars need to be designed that are outside the validity of these design guidelines, the advice of rock engineering specialists should be sought.” (NIOSH, 2011)

Partially benched pillar that failed along two angular discontinuities. Width-to-height ratio is 0.58 based on full benching height and average pillar stress is about 4 percent of the UCS.

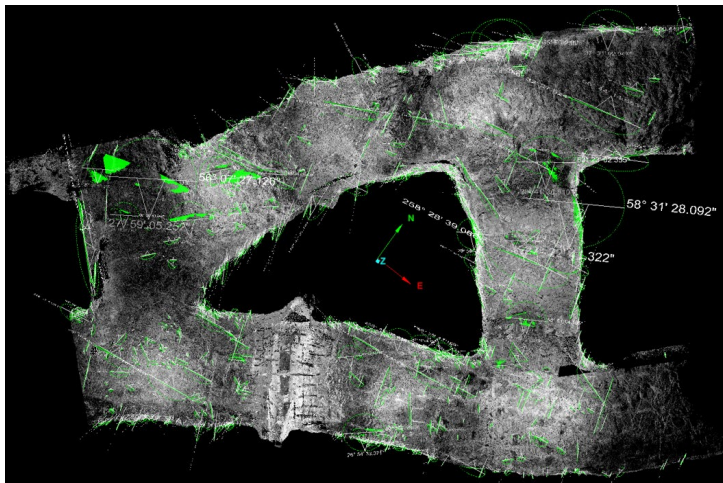


Pillar that has an original width-to-height ratio of 1.7 failed by progressive spalling. Thin weak beds are thought to have contributed to the failure. The average pillar stress was about 11 percent of the UCS prior to failure.

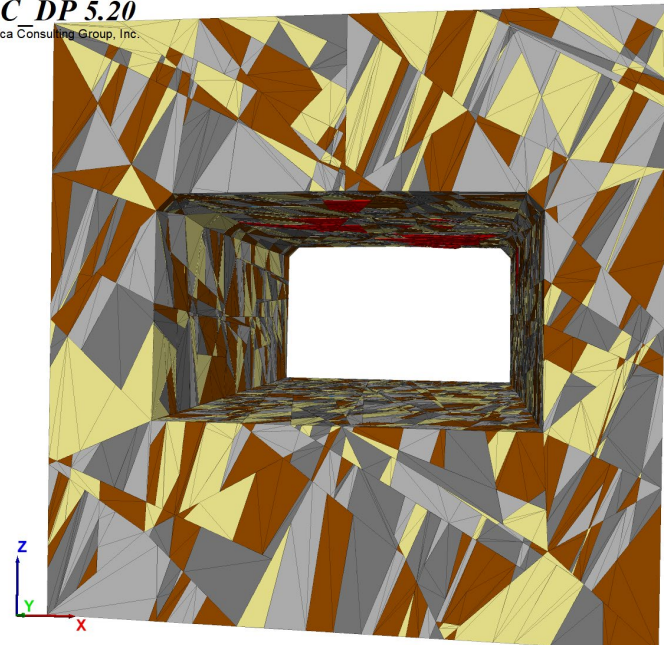


2. OBJECTIVE

To develop and implement a methodology that integrates laser scanning technology along with Discrete Element Modeling as tools for characterizing, preventing, and managing structurally controlled instability that may affect large-opening underground mines from a risk perspective.



3DEC DP 5.20
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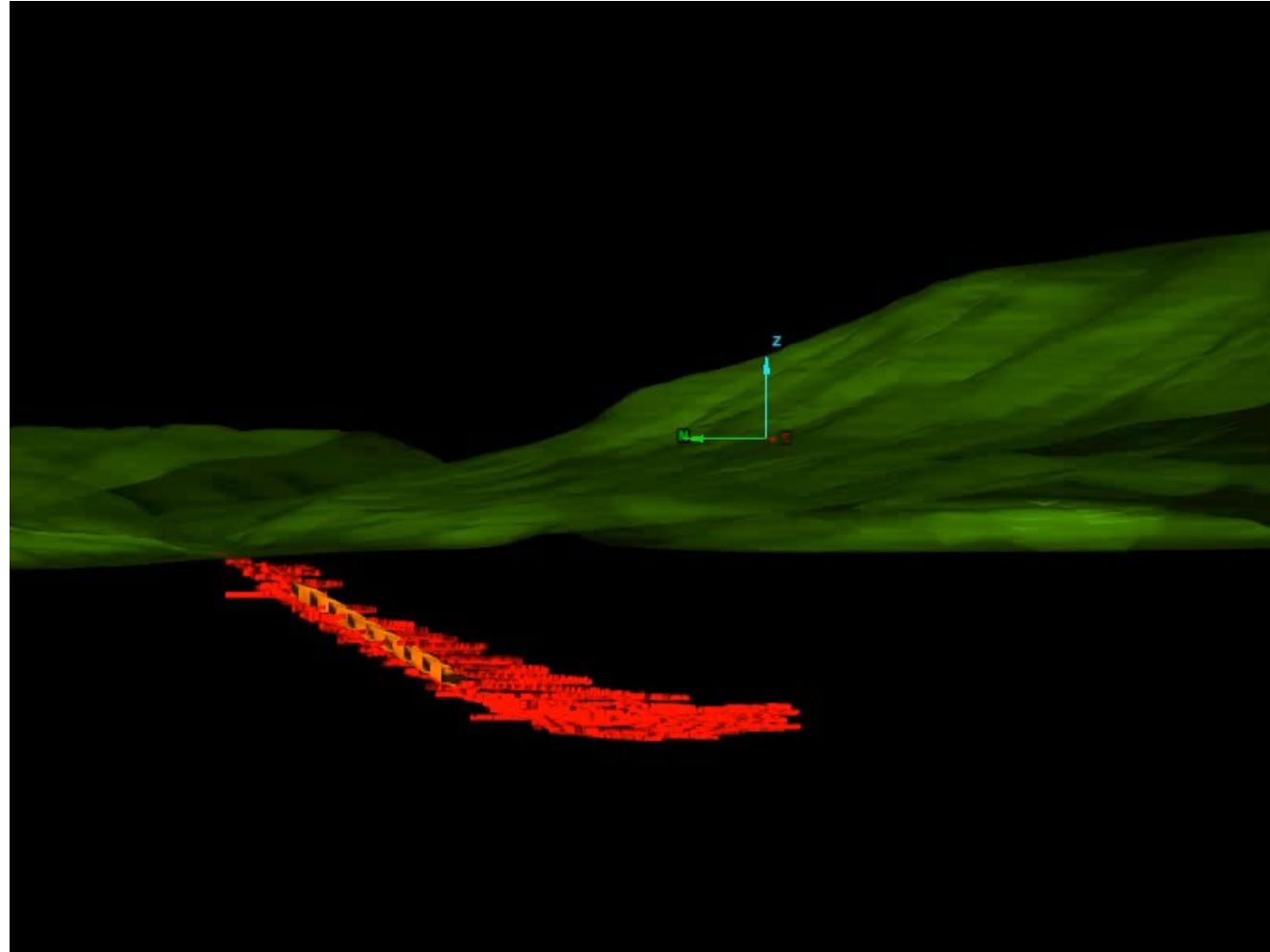




3. CASE STUDY MINE



3. CASE STUDY MINE



3. GEOTECHNICAL CONDITIONS



3. GEOTECHNICAL CONDITIONS

Description jointing pattern where at least four joint sets were well defined

observed wide-joint spacing, generally ranging from 0.6 m to 2 m,

amount of fallen blocks observed on the floor with cubical and tabular shapes,

joint surfaces defined as mostly closed, flat and smooth, with a JRC ranging from 2 to 4, completely dry and fresh,

other geological structures, such as faults and contacts, that could generate a rock fall in the absence of the required support

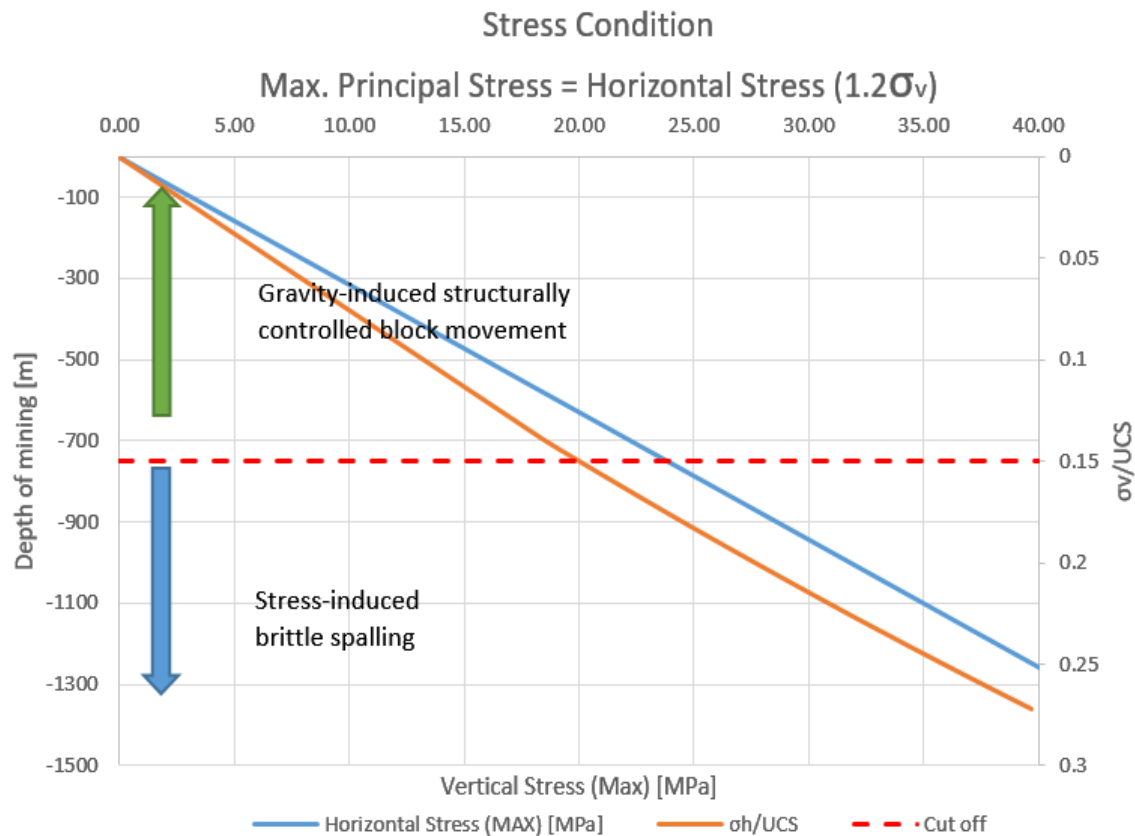


Rock Mass Classification

RMR	70 - 75	Good
Q	8 - 20	Fair
GSI	≈ 75	



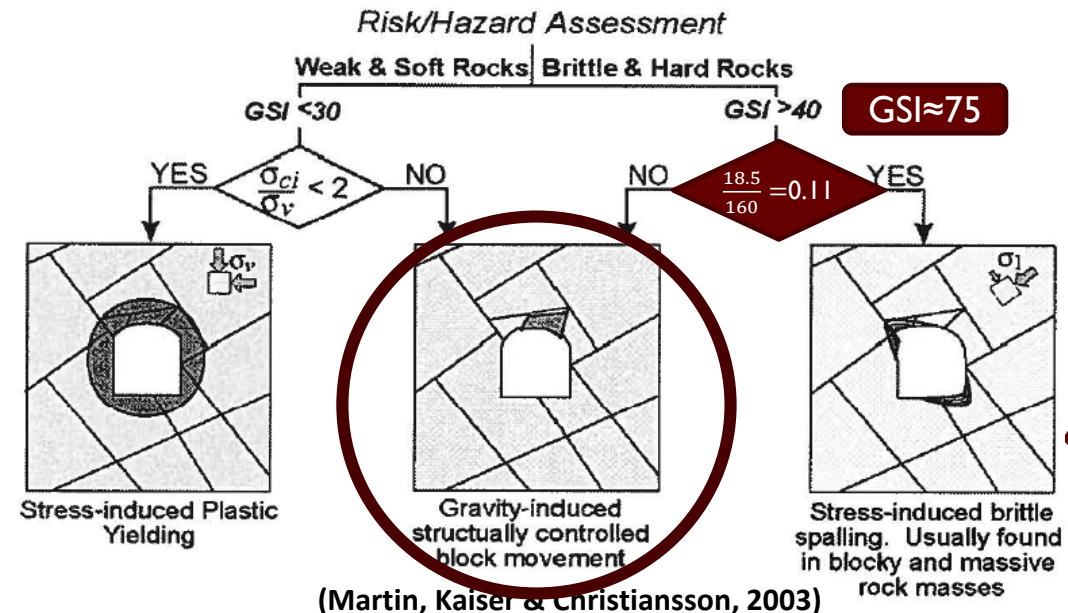
3. GEOTECHNICAL CONDITIONS



$$\sigma_v = \gamma h = 2.7 \frac{\text{ton}}{\text{m}^3} * 9.81 \frac{\text{m}}{\text{s}^2} * 700\text{m}$$

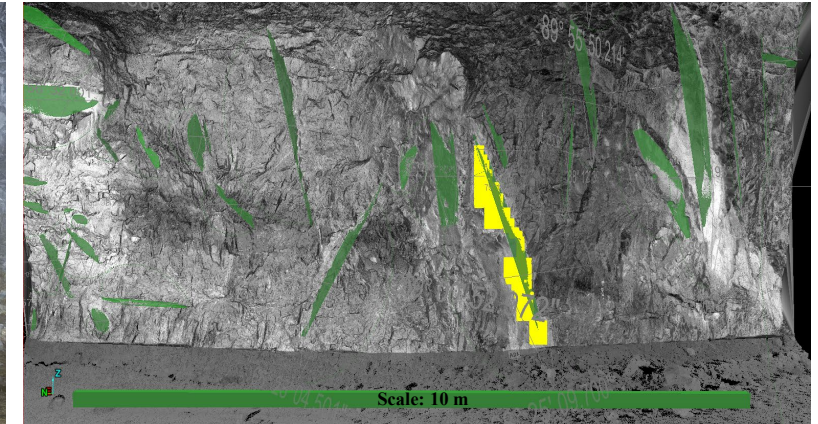
$$\sigma_v = 18.5 \text{ MPa}$$

Lithology	Density (ton/m ³)		UCS (MPa)		Brazilian Tensile Strength (MPa)		Young's Modulus (GPa)	
	Mean	SD	Mean	SD	Mean	SD	Mean	SD
Hanging Wall	2.69	0.01	163.74	37.84	11.96	3.14	61.02	6.79
Ore Body	2.69	0.01	159.20	21.25	6.30	1.99	64.11	2.37
Footwall	2.72	0.01	217.29	36.12	13.72	2.62	61.43	3.15



4. METHODOLOGY

1. Laser Scanning



2. Virtual Discontinuity Mapping

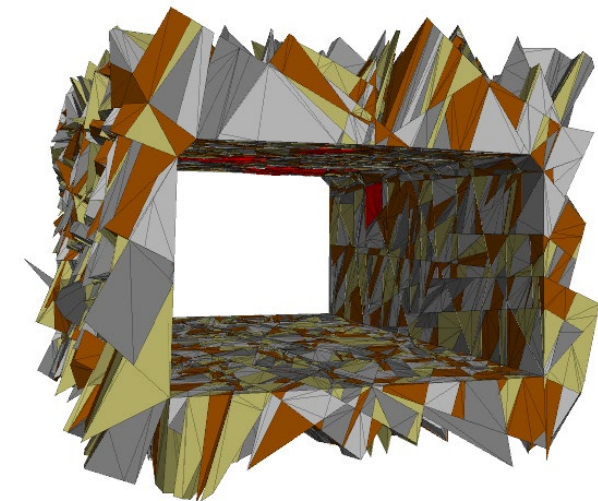
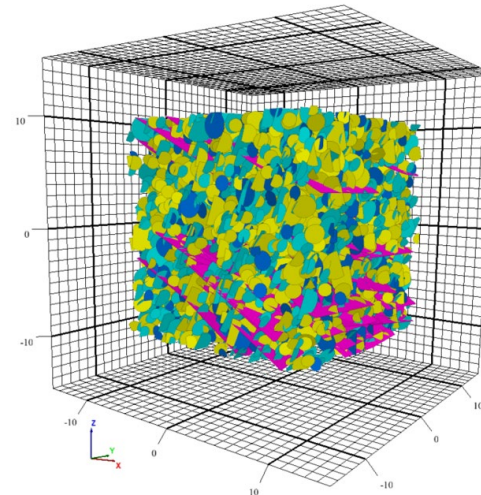


3. DFN Generation



4. Preliminary 3DEC Models

5. Stochastic 3DEC Modeling





5. RESULTS



5.1. LASER SCANNING - EQUIPMENT

FARO Focus 3D

Maximum Resolution 710.7 Million of points / scan

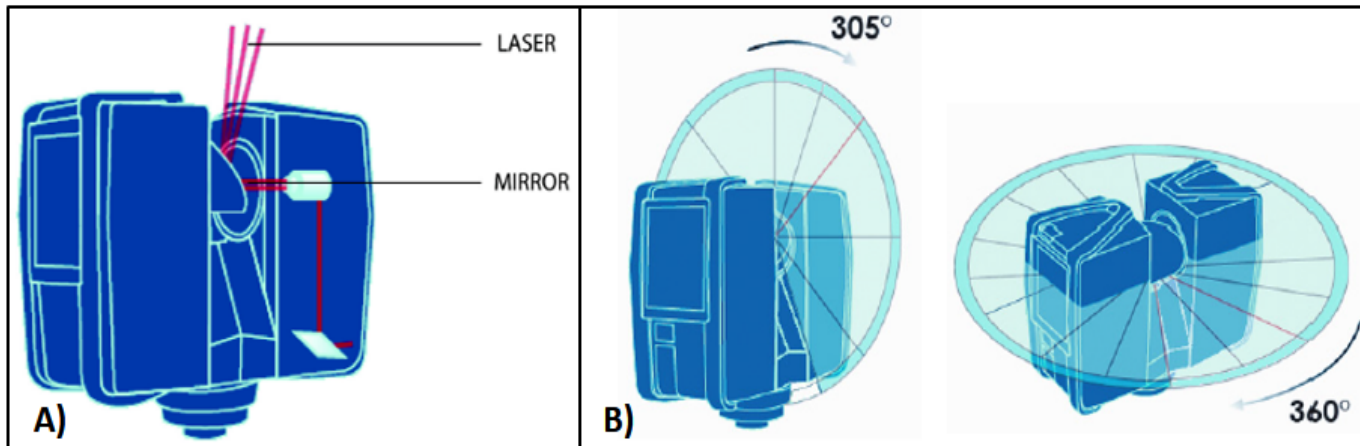
Purchased in 2011

Maximum Measurement Speed 976,000 pts/s

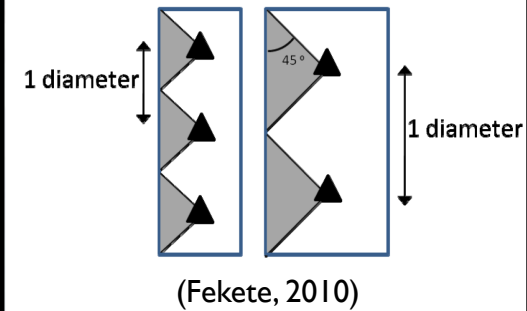
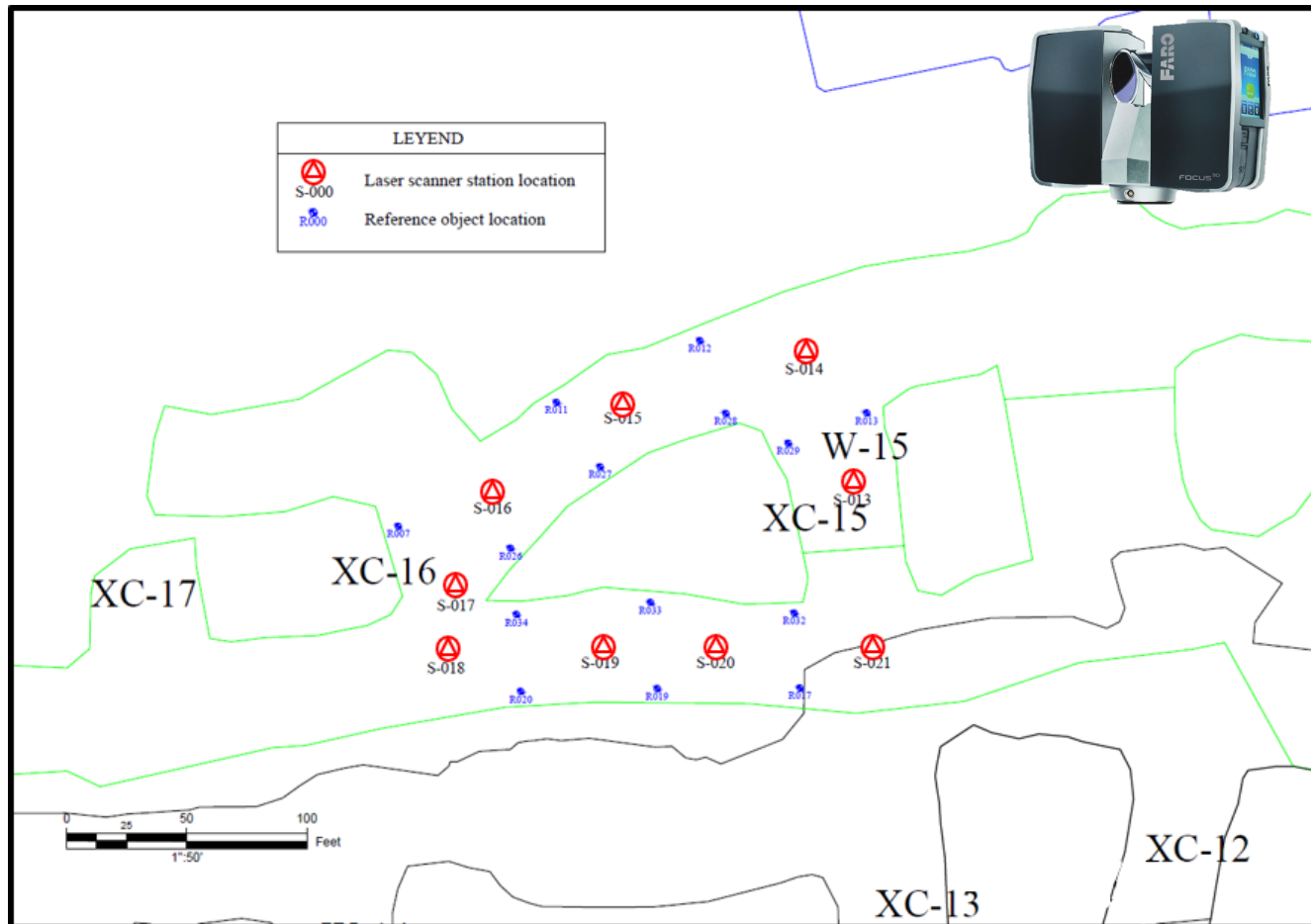
Ranging error $\pm 2\text{mm}$ @ 10 m

Internal Compass

No integrated GPS



5.1. LASER SCANNING – STATIONS LOCATION



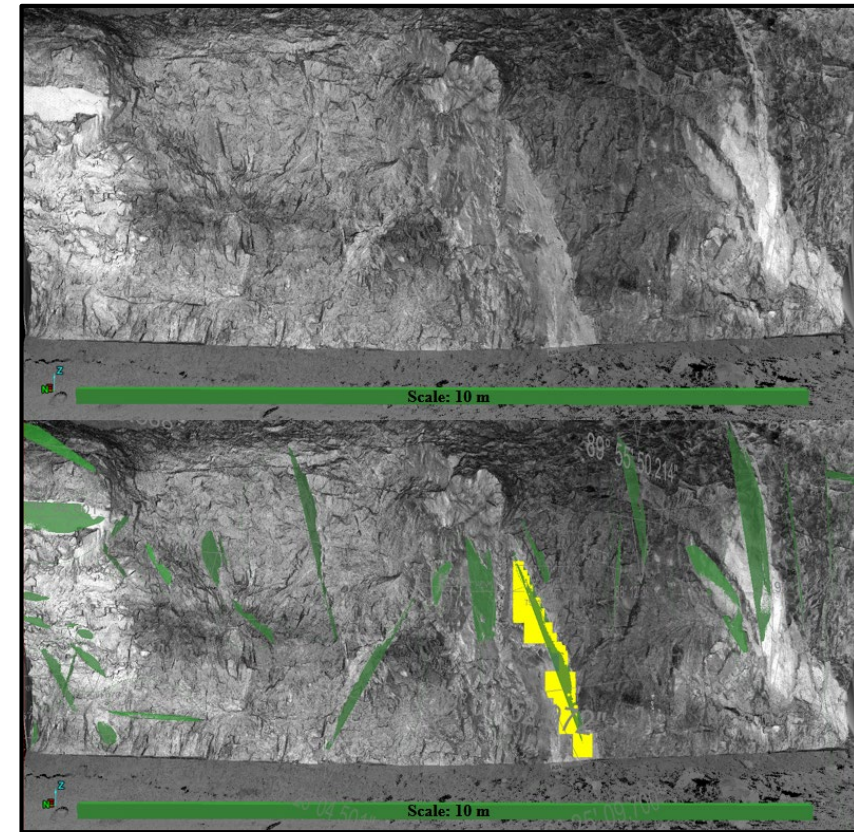
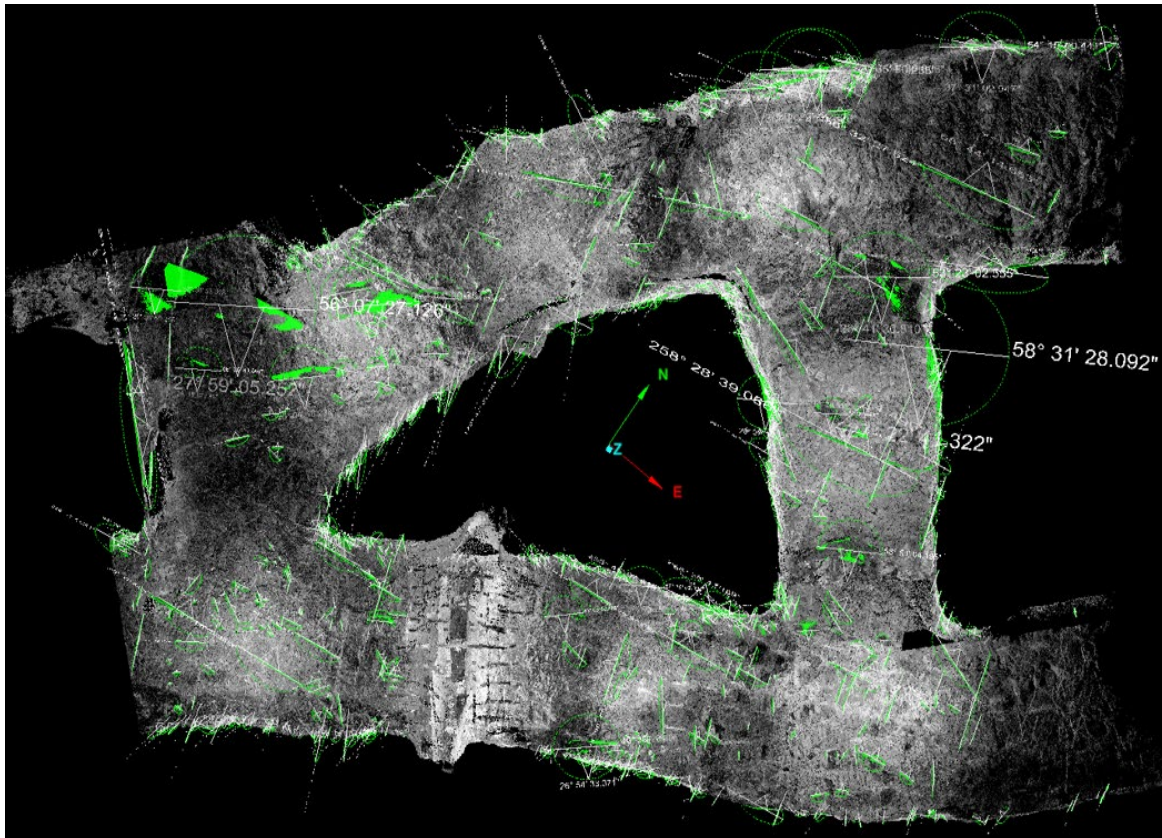
13 m Between stations

6.5 m Between stations
and reference objects

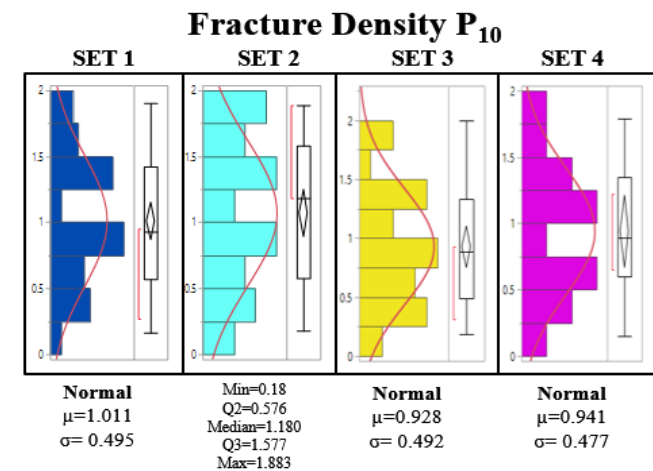
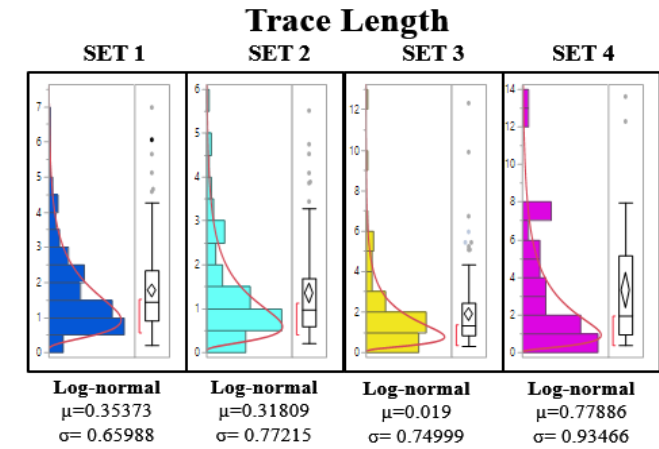




5.2.VIRTUAL DISCONTINUITY MAPPING



5.2.VIRTUAL DISCONTINUITY MAPPING - RESULTS



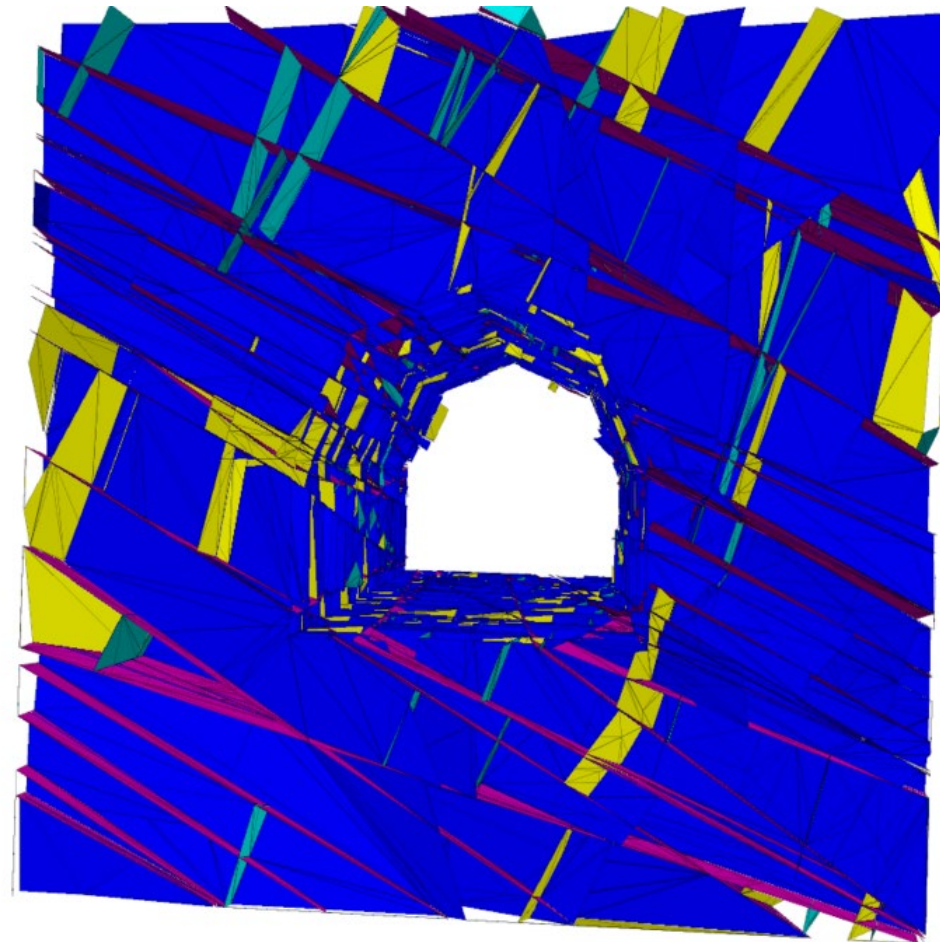
5.3. DISCRETE FRACTURE NETWORK (DFN) GENERATION

A DFN is a three-dimensional geometric representation of a geological structure based on statistical information of its characteristics measured on the field (Pierce, 2017).

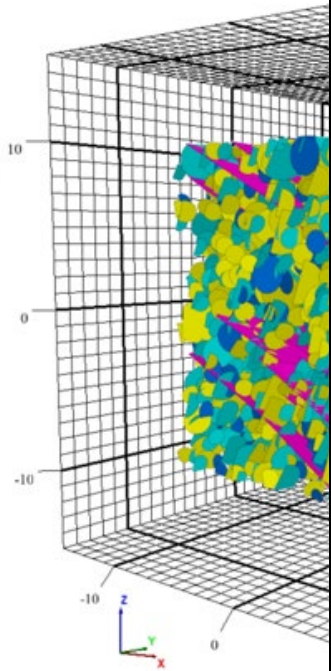
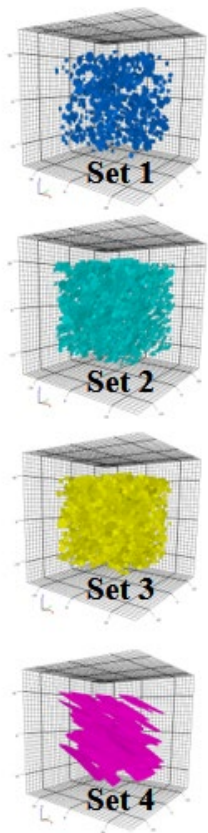
- Orientation
- Fracture Density
- Joint Size



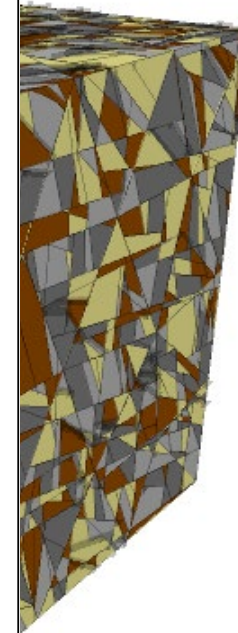
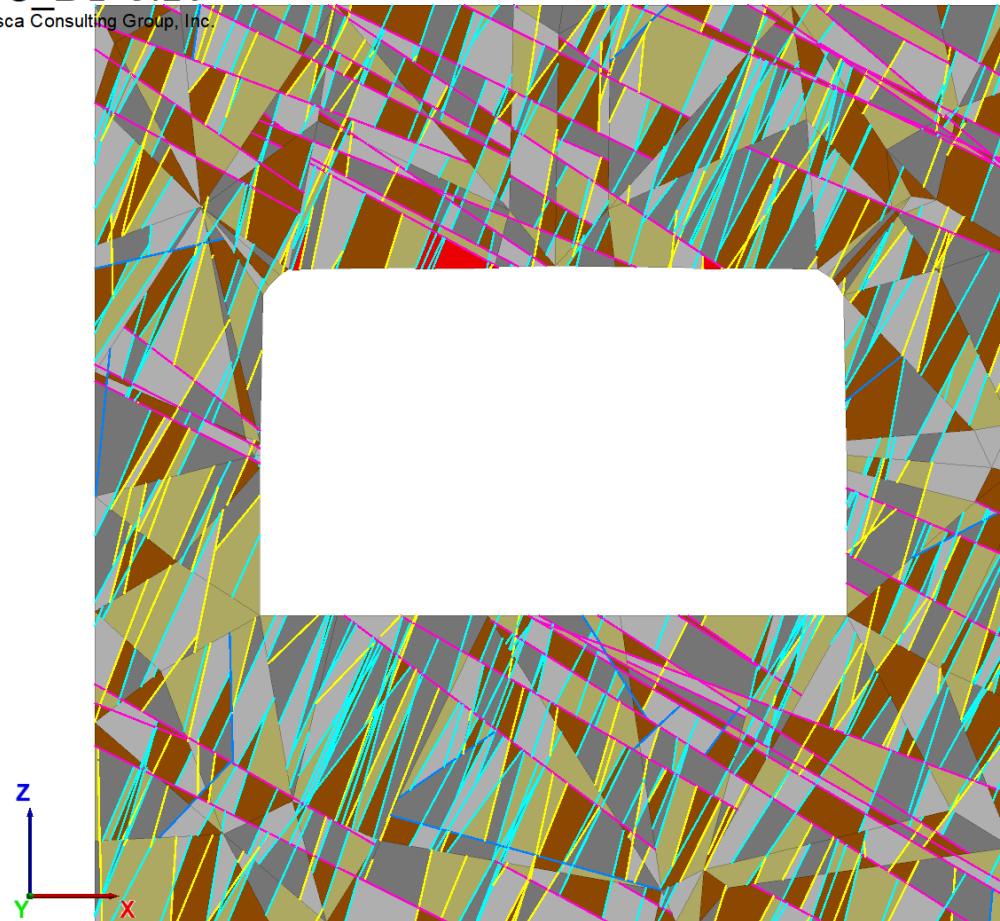
Exposure 4. Coastal cave in a layered rock mass, UK
(Hudson & Xia, 2015)



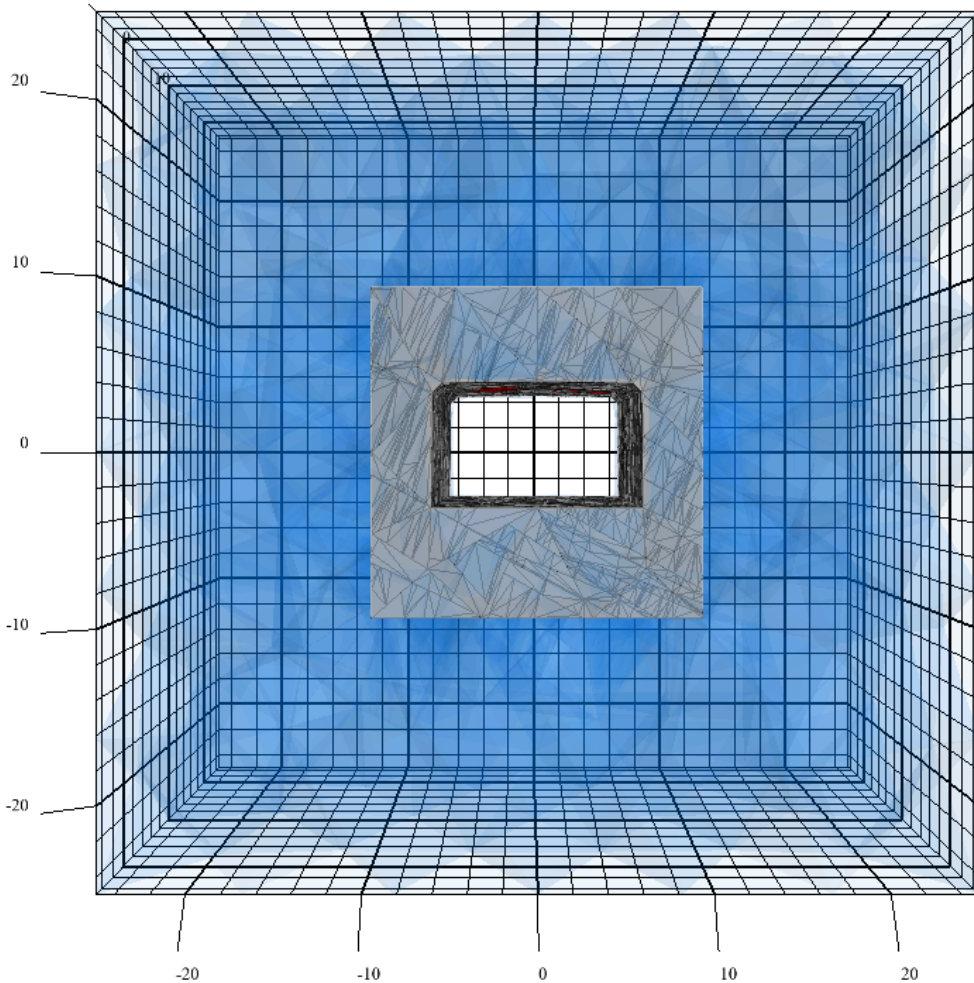
5.4. PRELIMINARY NUMERICAL MODEL (3DEC)



3DEC DP 5.20
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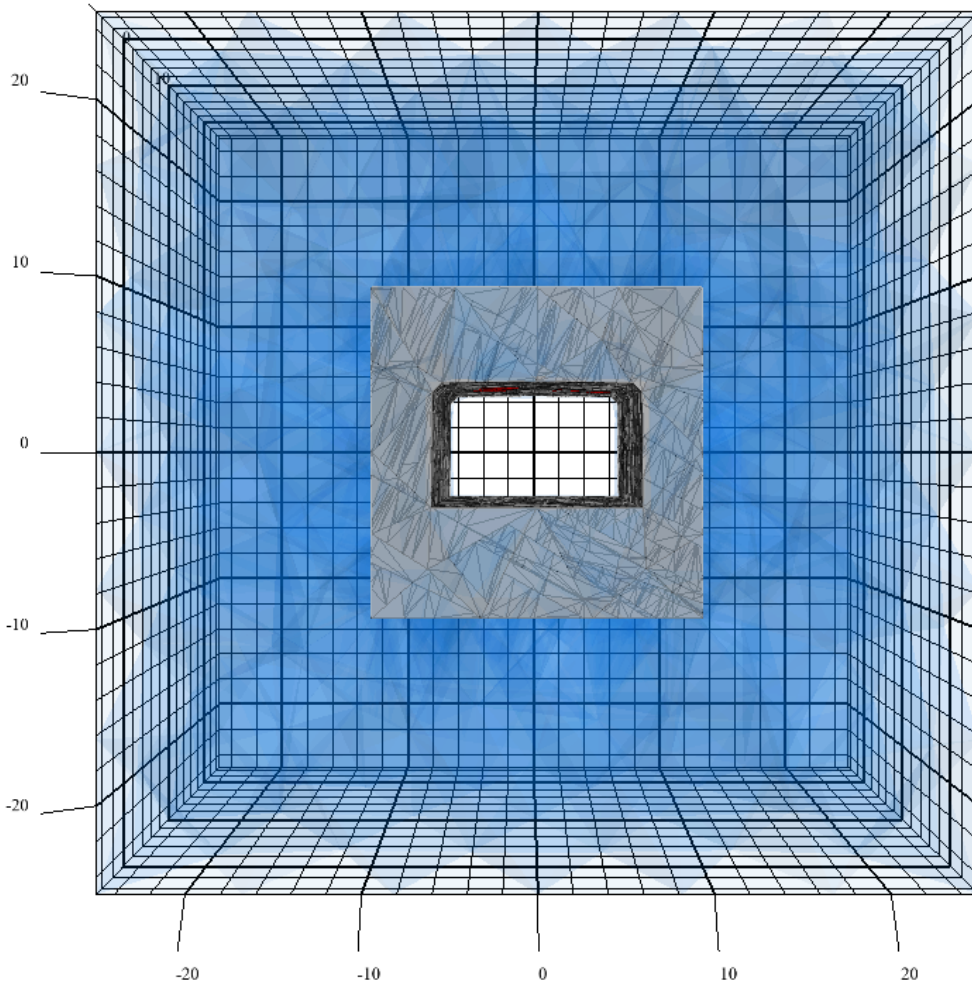
5.4. PRELIMINARY NUMERICAL MODEL (3DEC)



- Inner section: Fractured rock mass
 - Cut by 4 DFNs (20 m x 20 m x 20 m)
- External section: Massive rock mass – Fixed blocks
- Excavation Dimensions
 - 20 m length
 - 12.8 m width
 - 7.6 m height
- Stresses @ 700 m deep
 - $\sigma_v = 18.54 \text{ MPa}$
 - $\sigma_h = 20.39 \text{ MPa}$



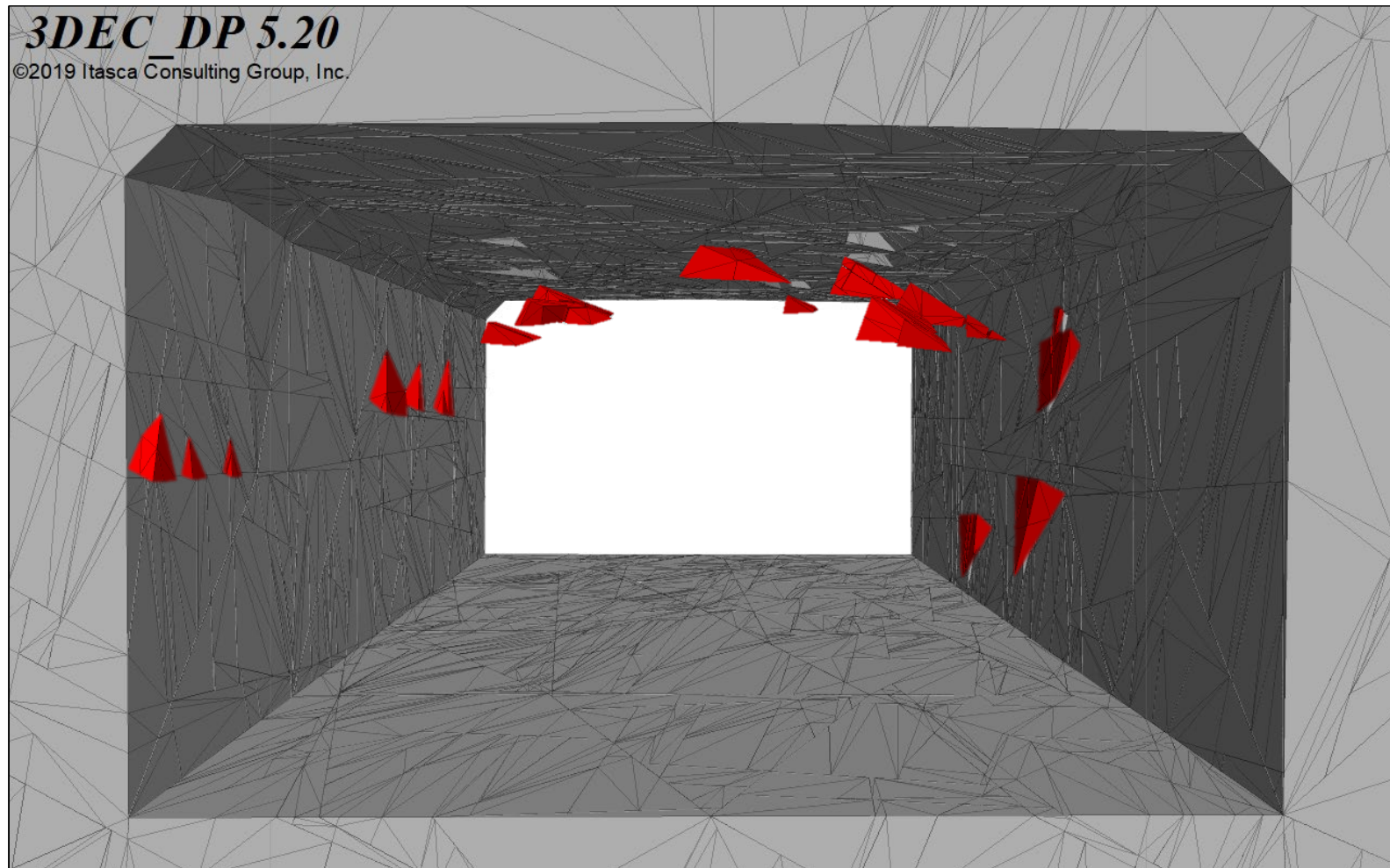
5.4. PRELIMINARY NUMERICAL MODEL (3DEC)



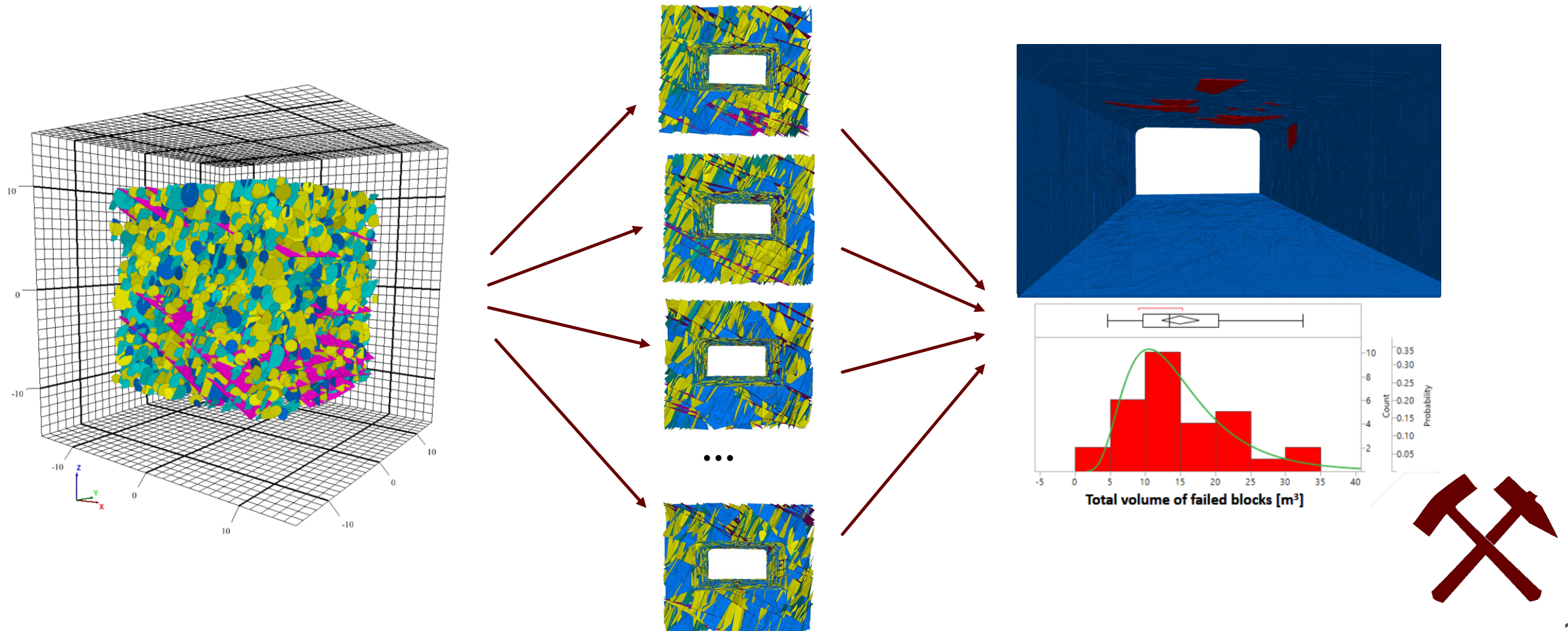
- Rigid block model (Blocks assumed infinitely stiff)
- Rock Density = 2.7 ton/m^3
- Mohr - Coulomb Constitutive Model for Fractures
 - $\phi = 30^\circ$ & $\text{cohesion} = 0$
 - $J_{ks} = 30 \text{ MPa/mm}$
 - $J_{kn} = 300 \text{ MPa/mm}$ (Bandis, Lumsden, & Barton, 1983)
- 1000 Initial cycles
- Progressive excavation (6 stages) – 5 m sections
 - 1000 cycles elastic – Damping
 - 1000 cycles plastic
 - time step of $3.81 \times 10^{-6} \text{ s}$



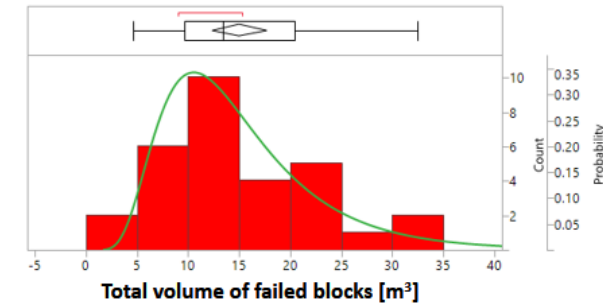
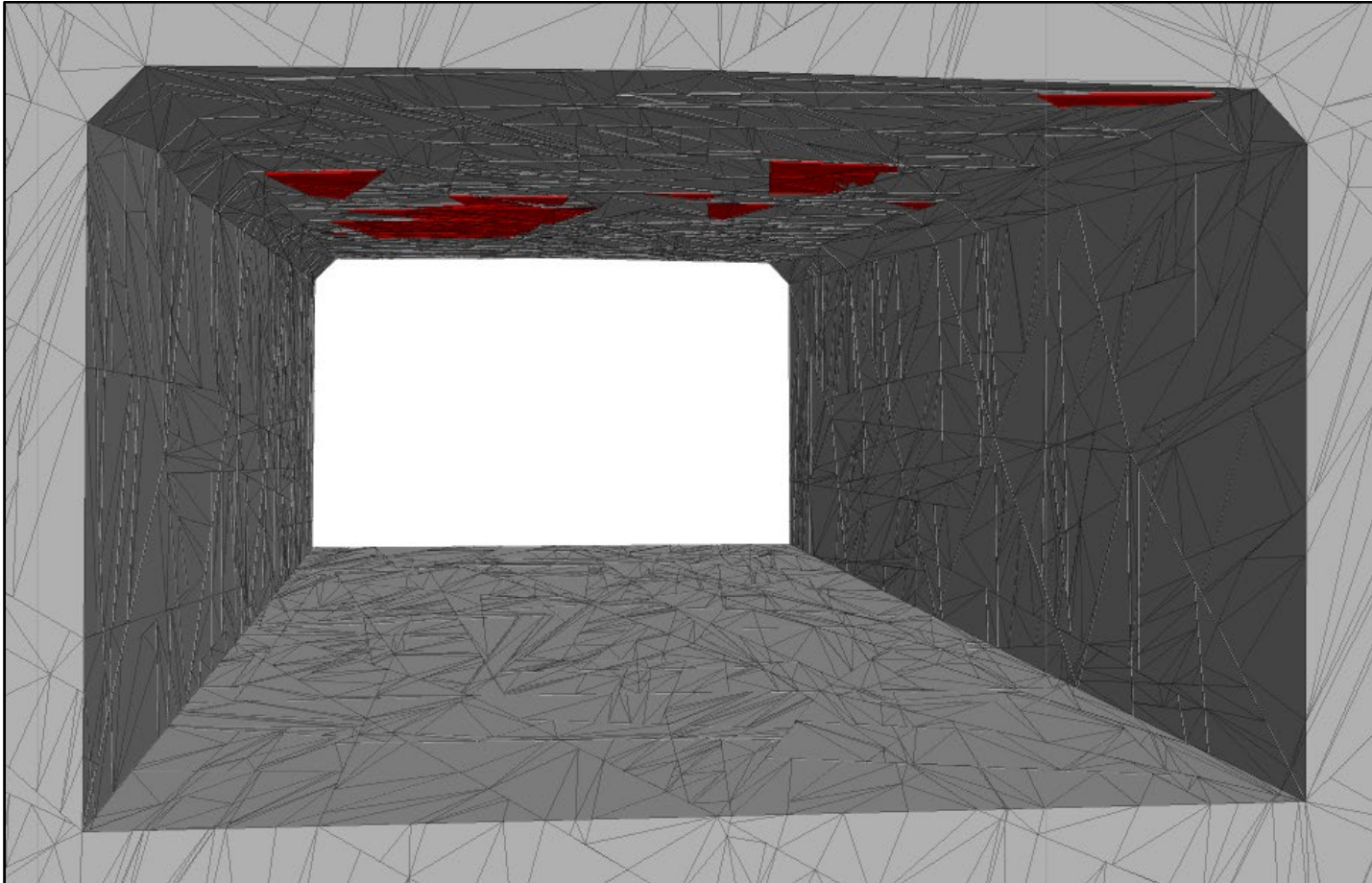
5.4. PRELIMINARY NUMERICAL MODEL (3DEC)



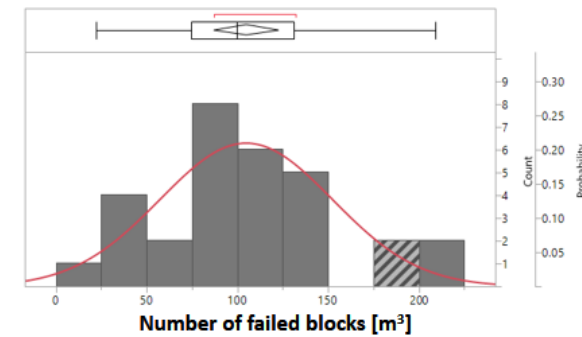
5.5. STOCHASTIC DISCRETE ELEMENT MODEL



5.5. STOCHASTIC DISCRETE ELEMENT MODEL - RESULTS



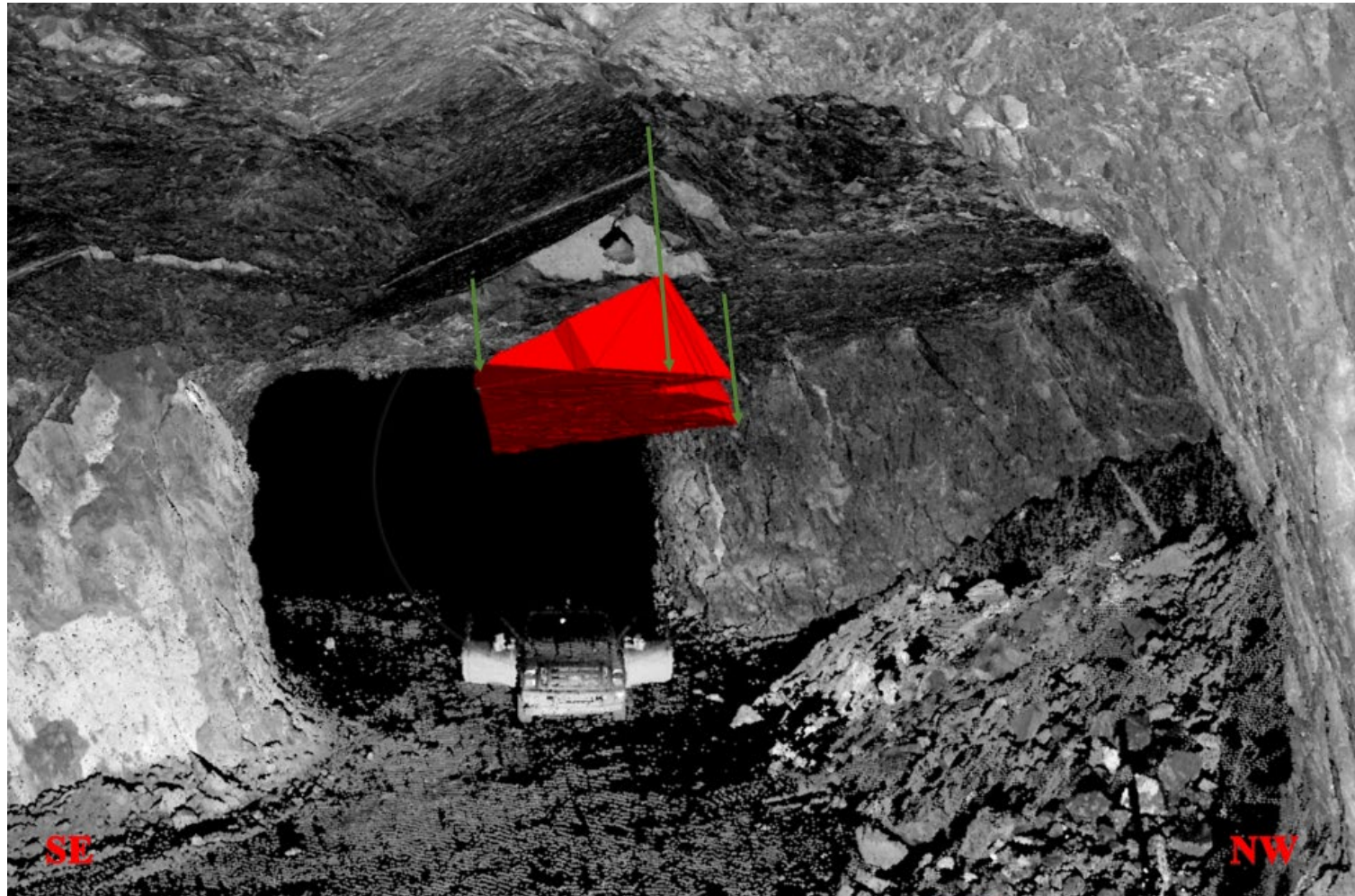
Summary Statistics	
Distribution	Log-Normal
Mean	2.60
Standard Deviation	0.49



Summary Statistics	
Distribution	Normal
Mean	104.87
Standard Deviation	47.58



5.6. MODEL VALIDATION



6. CONCLUSIONS

- TLS was proved to be a powerful tool for rock mass characterization.
- An adequate Laser scan project planning allowed
 - to save time during the scanning process
 - reduce errors on the resulting point clouds
 - obtain the necessary information required for virtual discontinuity mapping.
- DEM software such as 3DEC is a powerful tool to interpret and analyze structurally controlled instability in underground excavations.
- TLS and DEM were successfully integrated to estimate the probability of rock falls in an underground excavations.



6. CONCLUSIONS

- A clear understanding of the geological model of the site is important in order to improve the results from virtual discontinuity mapping and the fractured rock mass model.
- A rigid block model assumption yielded results that agreed with the field observations and laser scanning results.
- A stochastic Discrete element modeling approach was selected for the analysis due to the stochastic nature of DFNs.
- The present methodology can be used as a method for rock fall hazard identification in underground limestone mines. It can be easily integrated into a groundcontrol risk management system.
- The case study mine offered an ideal environment to apply both technologies.



QUESTIONS?



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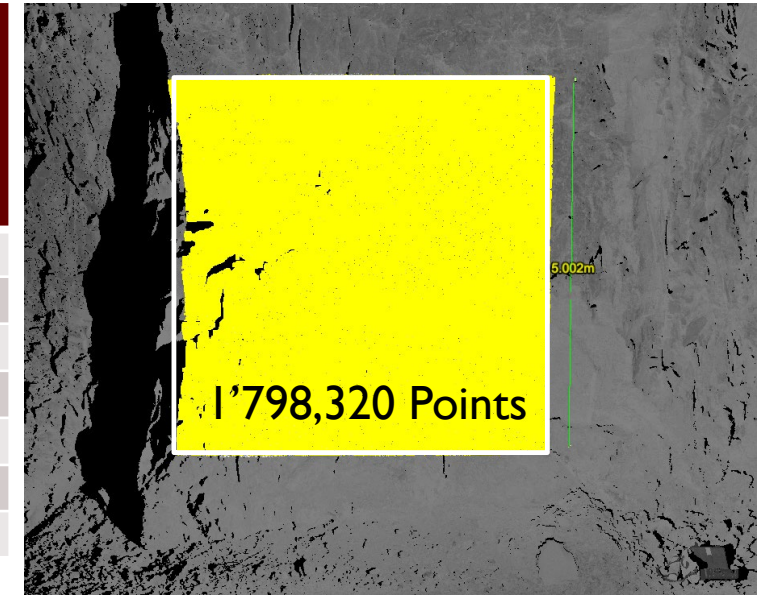
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5.1. LASER SCANNING - OPERATIONAL CONDITIONS

Scan	Resolution		Quality	Real Scan time [hh:mm:ss]	Point Count	Average Point Cloud Density [Points/cm ²]	Overall Rating
	Millions of Points/Scan						
S-006	44.4	1/4	1x	0:05:12	38,103,388	11.42	5.04
S-007	44.4	1/4	2x	0:06:06	41,229,797	12.48	4.31
S-008	44.4	1/4	3x	0:07:53	41,684,133	12.64	3.36
S-009	44.4	1/4	4x	0:11:28	41,229,797	12.66	2.34
S-010	28.4	1/5	2x	0:05:27	26,273,301	8.06	4.79
S-011	28.4	1/5	3x	0:06:36	26,573,225	8.13	3.96
S-012	28.4	1/5	4x	0:08:54	26,630,099	7.97	2.95

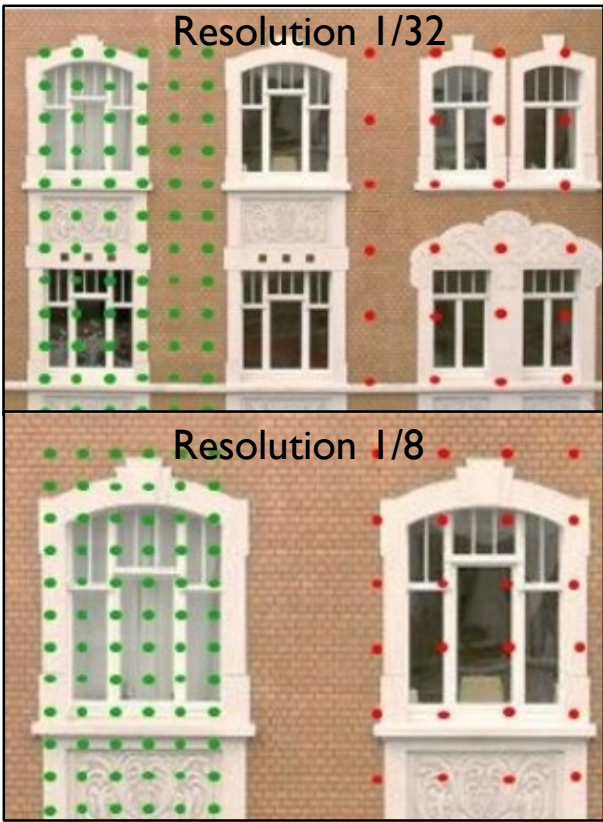


Point Cloud Densities for Structural Mapping

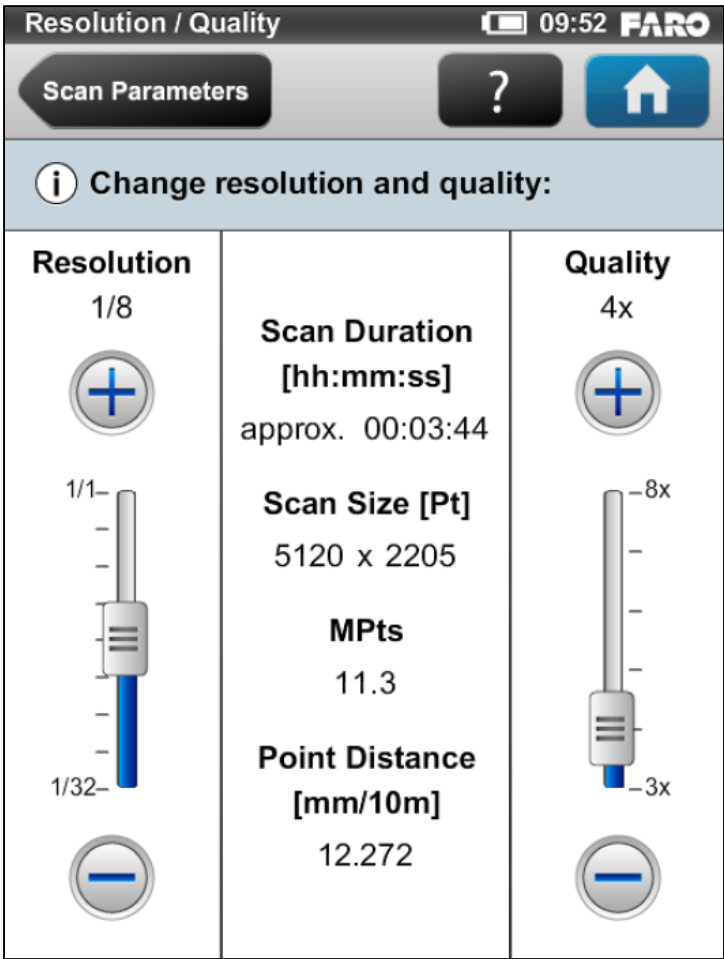
Author	Point cloud density
Lato, Diedrichs, Hutchinson and Harrap, 2009	4 points/cm ²
Cacciari & Futai, 2017	16 points/cm ²
Monsalve, Baggett, Bishop & Ripepi, 2018	11 points/cm ²



5.1. LASER SCANNING - OPERATIONAL CONDITIONS



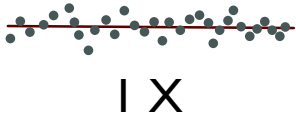
- Laser scanner at a distance of 10m
- Laser scanner at a distance of 20m



Quality	Observation Time
1 X	1 μ s
4 X	8 μ s

- Increasing the quality reduces the noise in the distance measurement.

✓ Thickness of a flat object



1 X

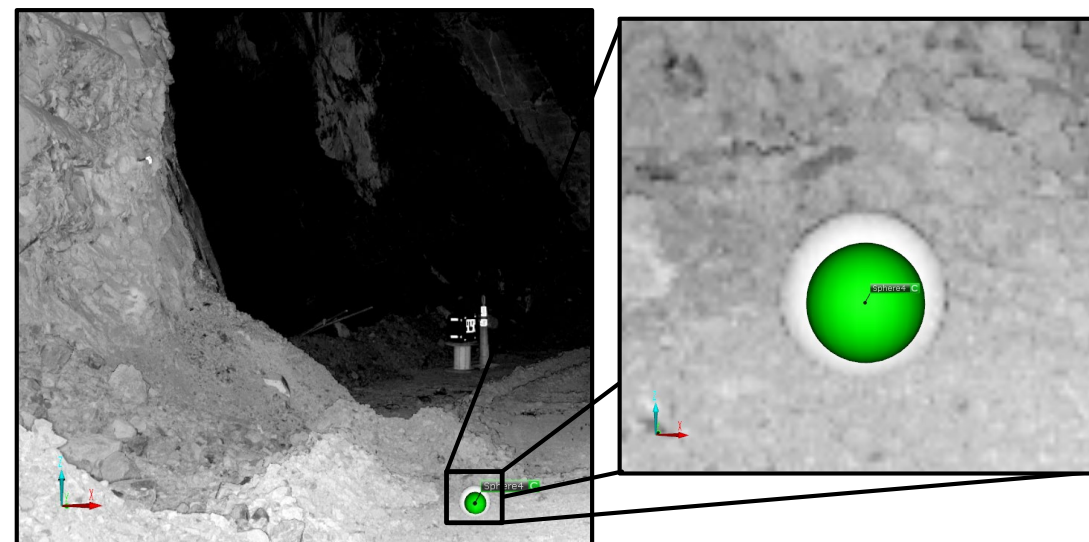
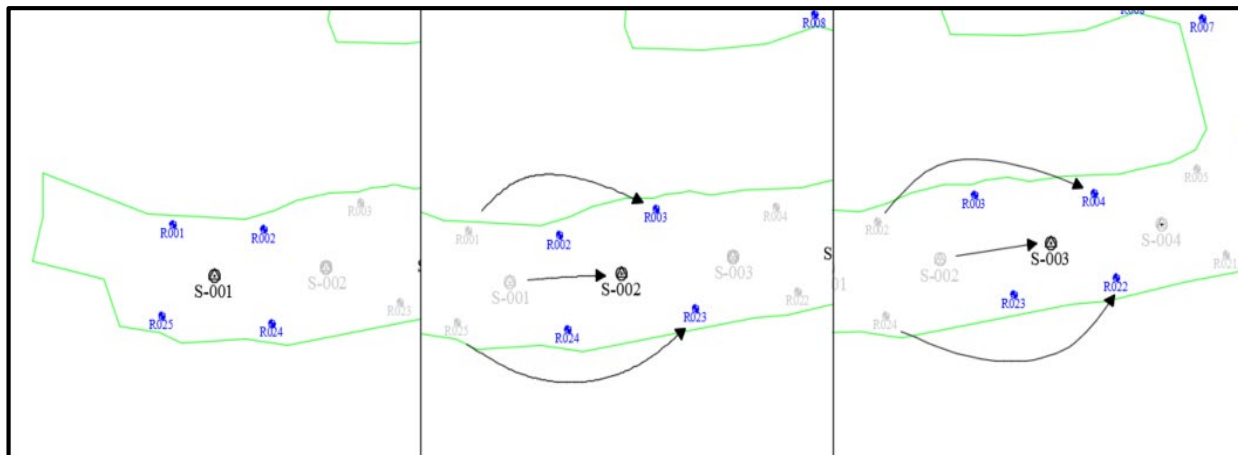


4 X

✓ Scan points in far objects

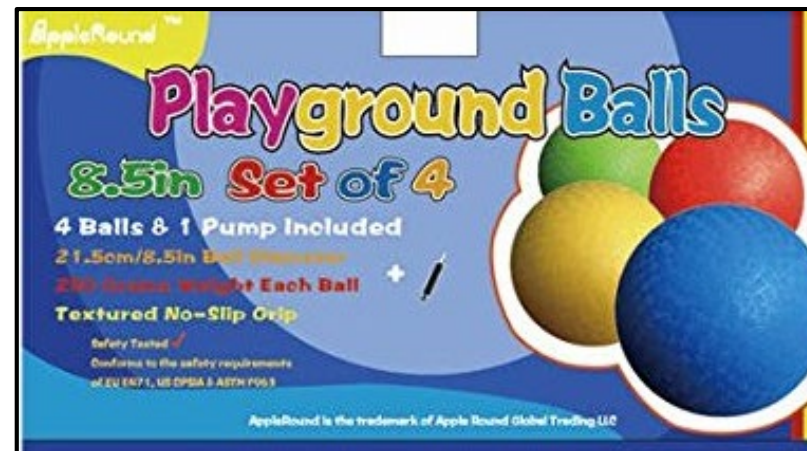


5.1. LASER SCANNING - SCANS REFERENCING

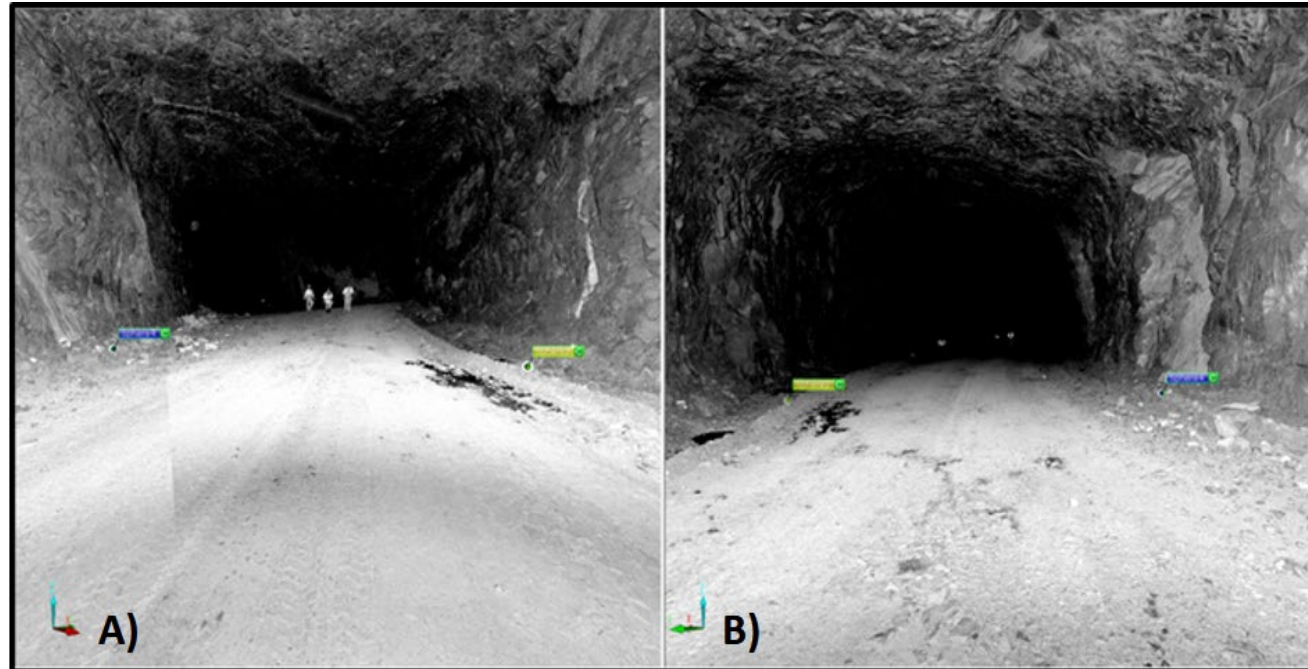


Recommended target spacing

Sphere size	145 mm	Sphere size	230 mm
Resolution Setting	Target Distance (max) in m	Resolution Setting	Target Distance (max) in m
1/16	5	1/16	7
1/10	7	1/10	11
1/8	9	1/8	14
1/5	15	1/5	22
1/4	18	1/4	27
1/2	37	1/2	55
1/1	73	1/1	110



5.1. LASER SCANNING - DATA REGISTRATION

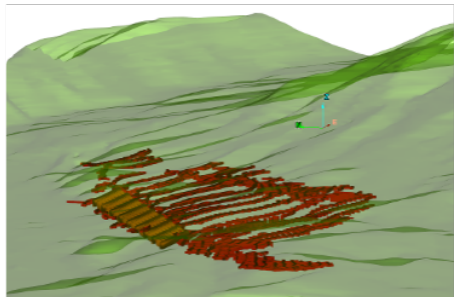
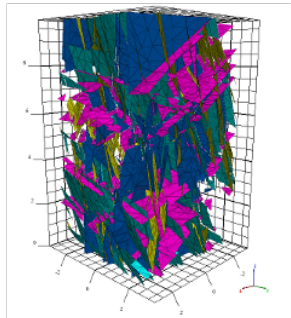
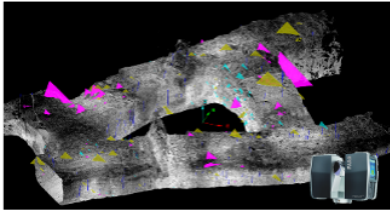


Scan Point Statistics	Obtained Values	Acceptable values	Unacceptable values
Maximum Point Error	6.7 mm	< 8 mm	> 20 mm
Mean Point Error	4.3 mm		
Minimum Overlap	25.2 %	> 25.0 %	< 10.0 %

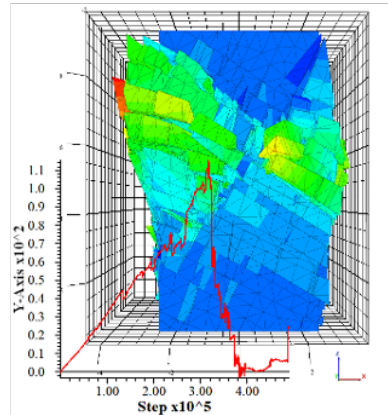


8. FUTURE WORK

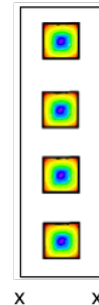
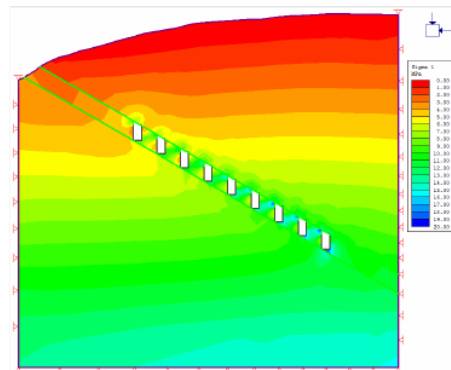
Site Scanning and Discontinuity Characterization



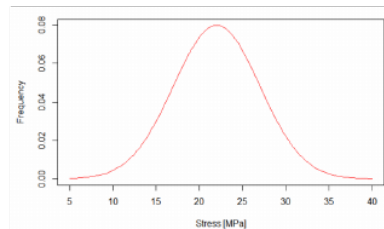
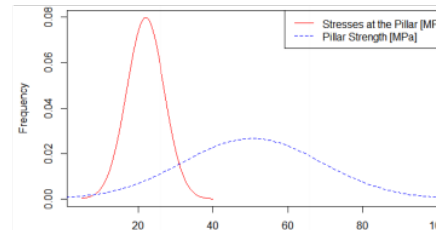
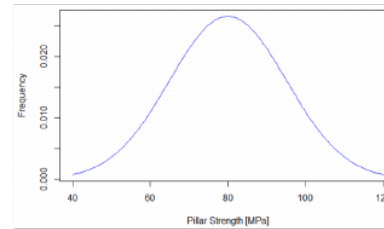
Pillar Strength Determination



Stress level Estimation



Stochastic Analysis and Probability Distribution Estimation



Probability of Failure Estimation

Factor of safety:

$$F.S. = \frac{\text{Strength}}{\text{Stress}} = \frac{S_p}{\sigma_p}$$

Limit state function:

$$g(\sigma_p, S) = S_p - \sigma_p$$

Failure set:

$$P(\text{failure}) = P\{F.S. < 1\} = P\{S_p < \sigma_p\}$$

Reliability index:

$$\beta = \frac{\mu_g}{\sigma_g}$$

Expected value of the limit state function:

$$\mu_g = \mu_{S_p} + \mu_{\sigma_p}$$

Standard deviation of the limit state function:

$$\sigma_g = \sqrt{\sigma_{S_p}^2 + \sigma_{\sigma_p}^2 - 2\rho_{S_p\sigma_p}\sigma_{S_p}\sigma_{\sigma_p}}$$

Pillar probability of failure:

$$Pf = 1 - \Phi(\beta)$$



