

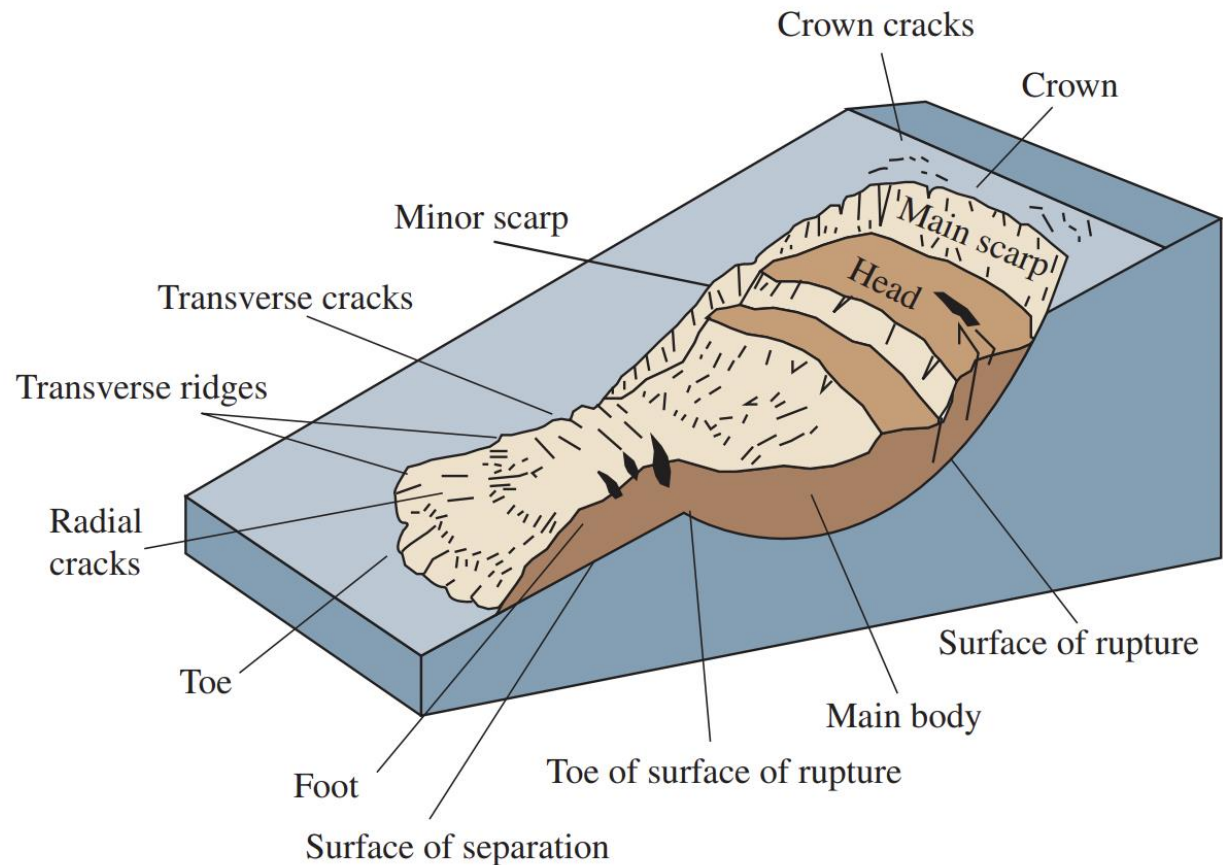


GEO-ENGINEERING TECHNIQUES FOR UNSTABLE SLOPES

Group 4 Presentation



- What is a “Landslide”?
- Landslide Types
- Landslide Parts



Comparison Between Experiment 0 and 1

- Experiment 0

Front view

5 cm
5 cm
5 cm



80 cm

Uniformly
compacted
sand

Side view



200 cm

- Experiment 1

Front view

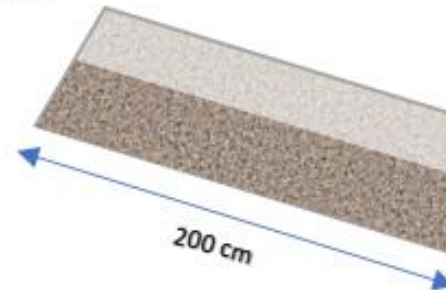
7.5 cm
7.5 cm



80 cm

uncompact sand
compacted sand

Side view

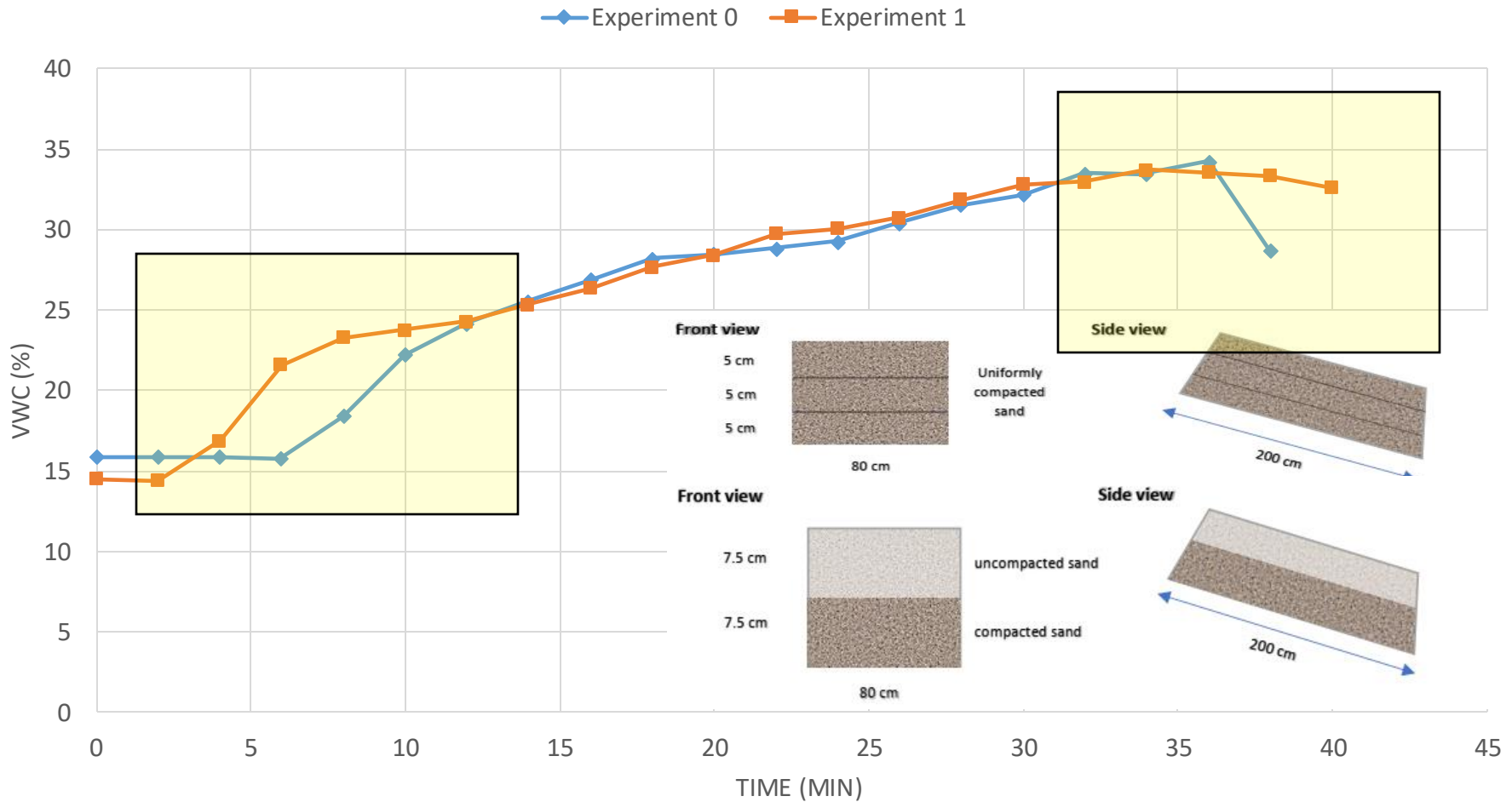


200 cm



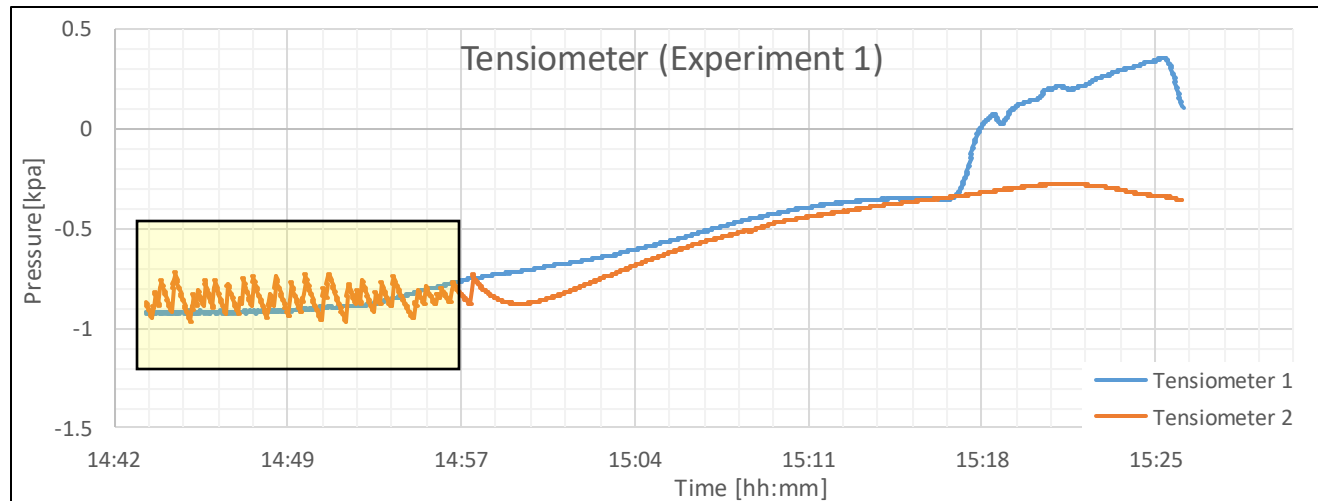
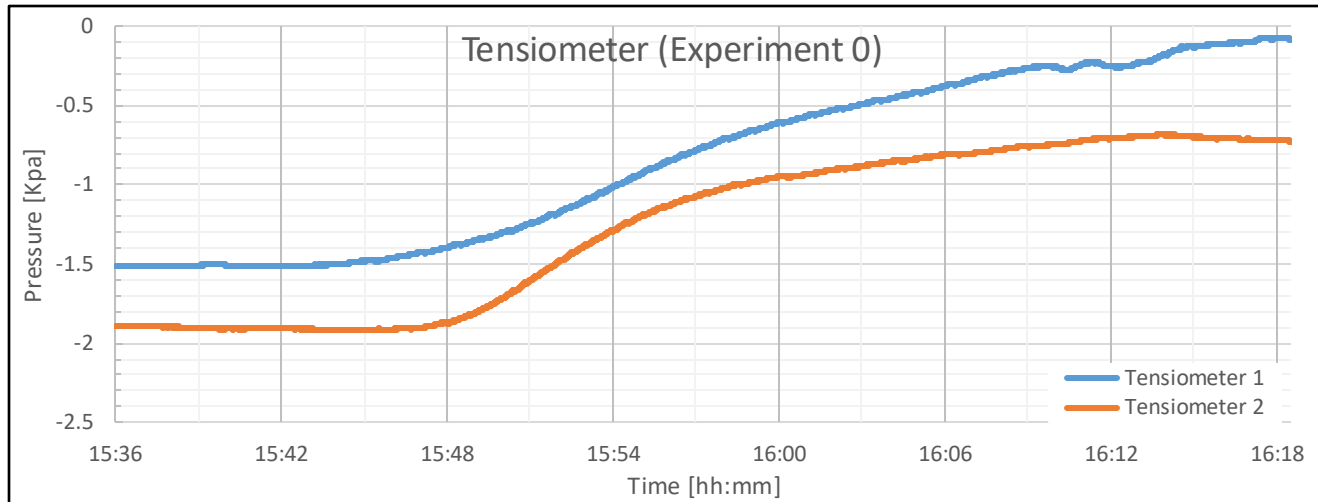
Comparison Between Experiment 0 and 1

TDR RESULTS



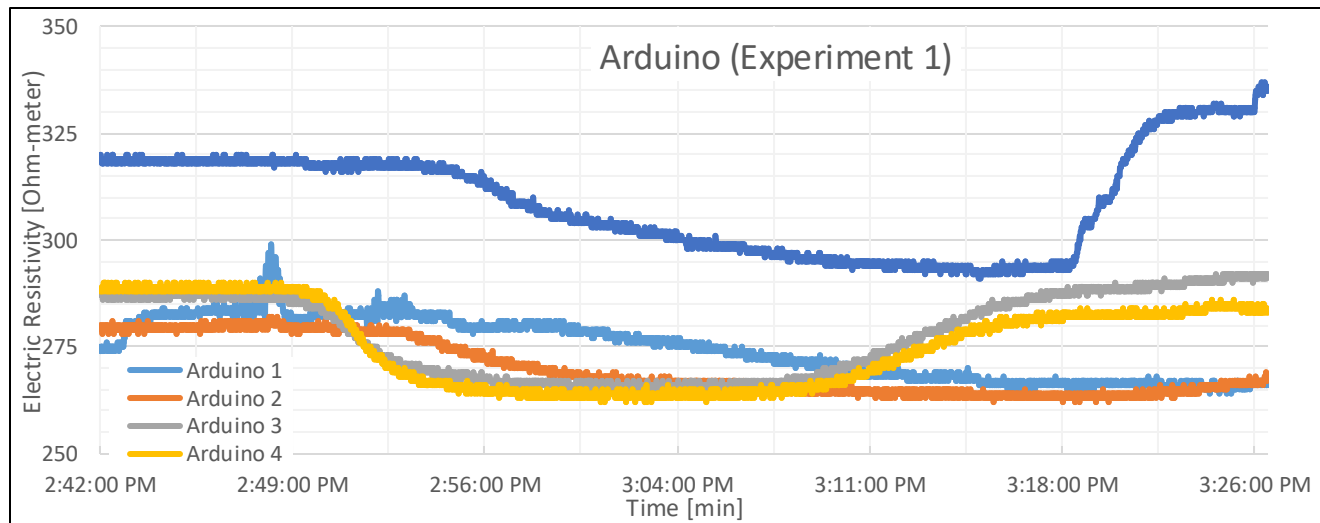
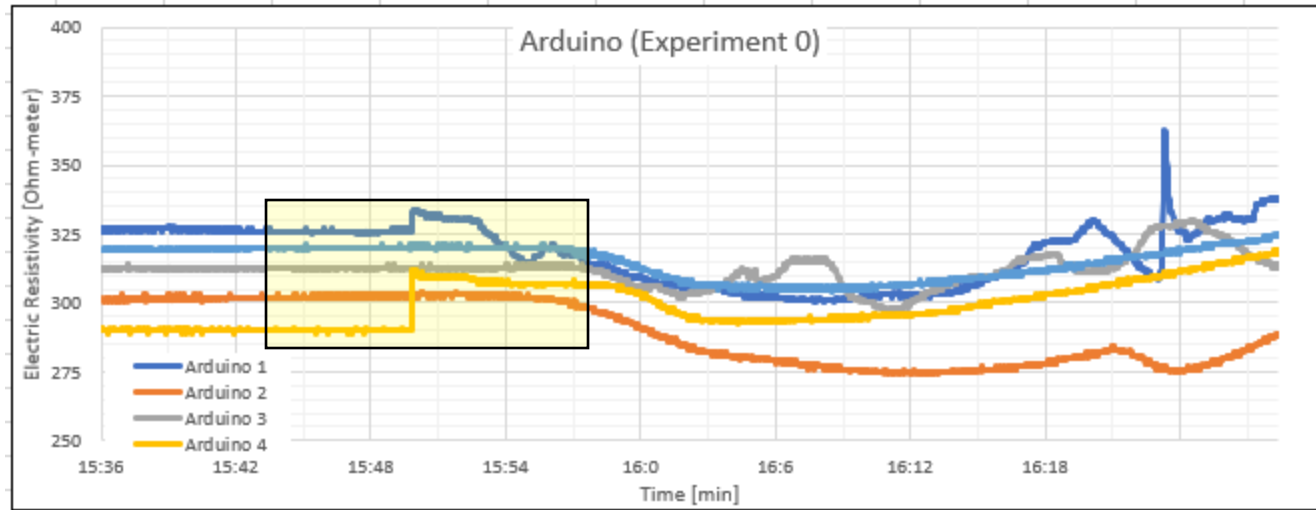


Comparison Between Experiment 0 and 1





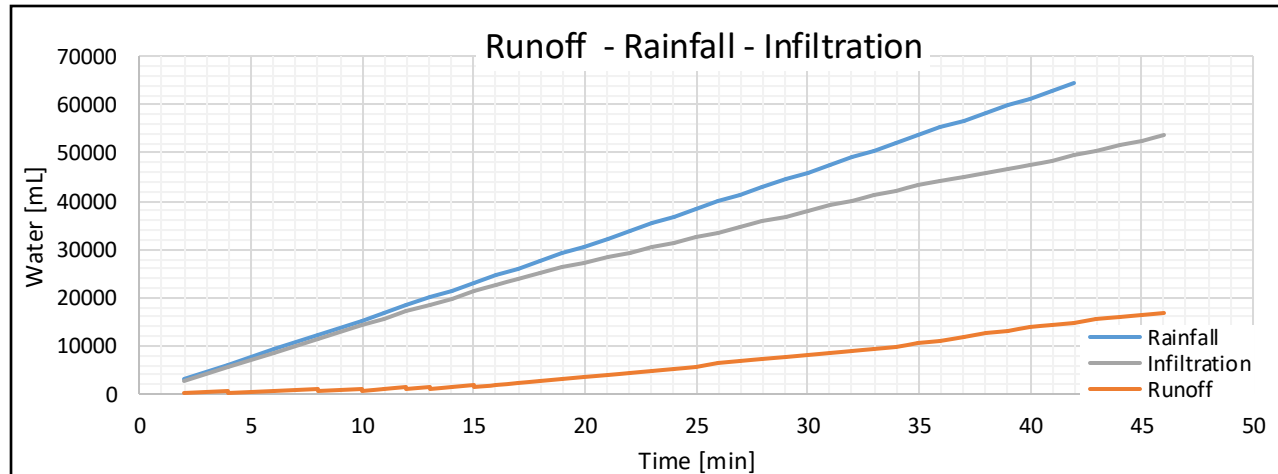
Comparison Between Experiment 0 and 1





Comparison Between Experiment 0 and 1

	Area [m ²]	1.6				
	N sprinklers	4	6	4	6	
Pressure	Discharge	Intensity [mm/h]		Intensity [m/min]		
0.5	0.22	33.0	49.5	0.000549938	0.000824906	
0.6	0.24	35.9	53.8	0.00059828	0.00089742	
0.7	0.26	38.7	58.0	0.000644958	0.000967436	
0.8	0.28	41.4	62.1	0.00068997	0.001034955	
0.9	0.29	44.0	66.0	0.000733318	0.001099976	
1	0.31	46.5	69.8	0.000775	0.0011625	
1.1	0.33	48.9	73.4	0.000815018	0.001222526	
1.2	0.34	51.2	76.8	0.00085337	0.001280055	
1.3	0.36	53.4	80.1	0.000890058	0.001335086	
1.4	0.37	55.5	83.3	0.00092508	0.00138762	
1.5	0.38	57.5	86.3	0.000958438	0.001437656	
1.6	0.40	59.4	89.1	0.00099013	0.001485195	
1.7	0.41	61.2	91.8	0.001020158	0.001530236	
1.8	0.42	62.9	94.4	0.00104852	0.00157278	





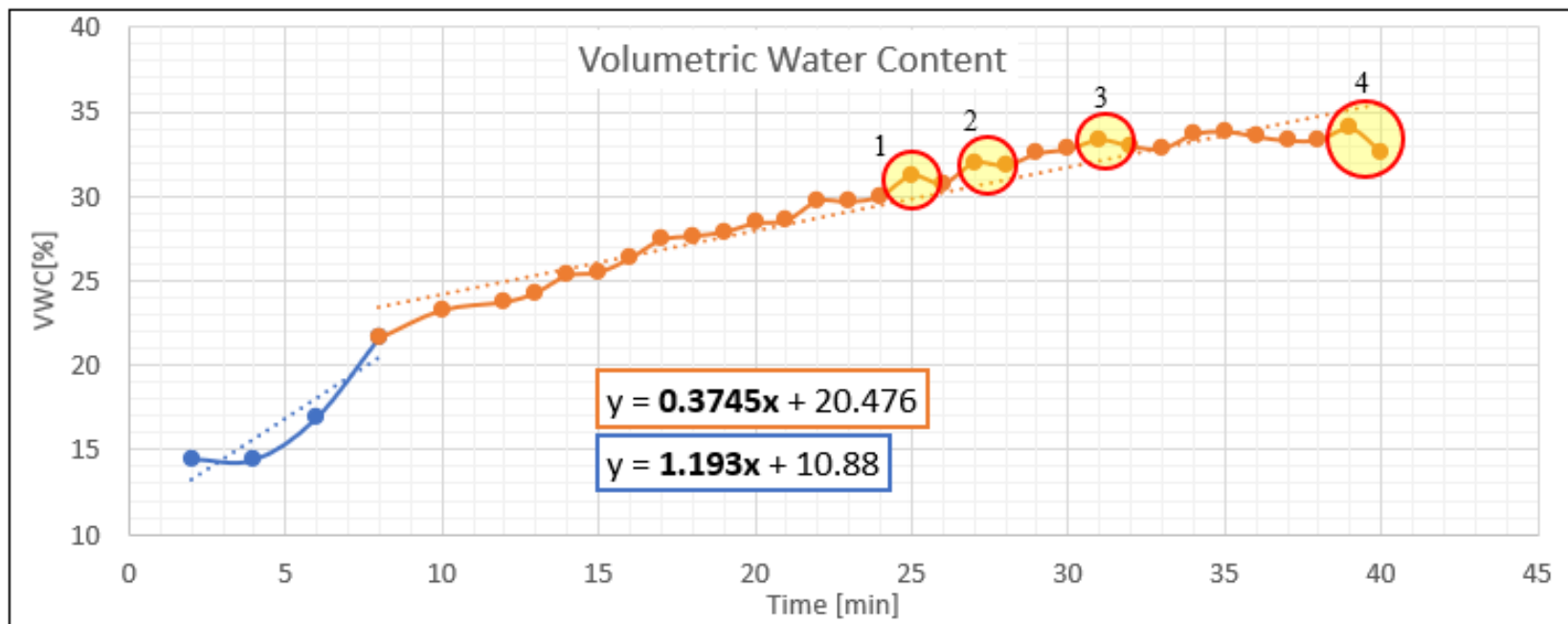
List of Anomalies

Table 3. List of Anomalies and Related Information

Geophysical Assessment			Photogrammetry Assessment		Geophysical Assessment			
Anomaly No.	Exp. Equipment	Time	Fracture ID	Time	Measurement ID	Time		
1	TDR Exp.1	24-26	Fracture 2	24:10	T5	22		
					T6	27		
2	TDR Exp.1	27-29	Fracture 3	27:09	T6	27		
			Fracture 4	27:50				
			Fracture 5	28:20				
3	TDR Exp.1	31-33	Fracture 7	31:00	T7	33		
			Fracture 8	30:50				
			Fracture 9	38:10				
4	TDR Exp.1	38-40	Fracture 10	37:00	T9	37		
			Fracture 11	37:30				
			Fracture 12	38:00			T10	39
			Fracture 13	39:30				
			Fracture 14	30:45				
5	TDR Exp.0	35			T6	33		
6	Tensiometer Exp. 1	0.235			Same anomaly as No.4			
7	Arduino Exp. 1	15			Same anomaly as No. 4			

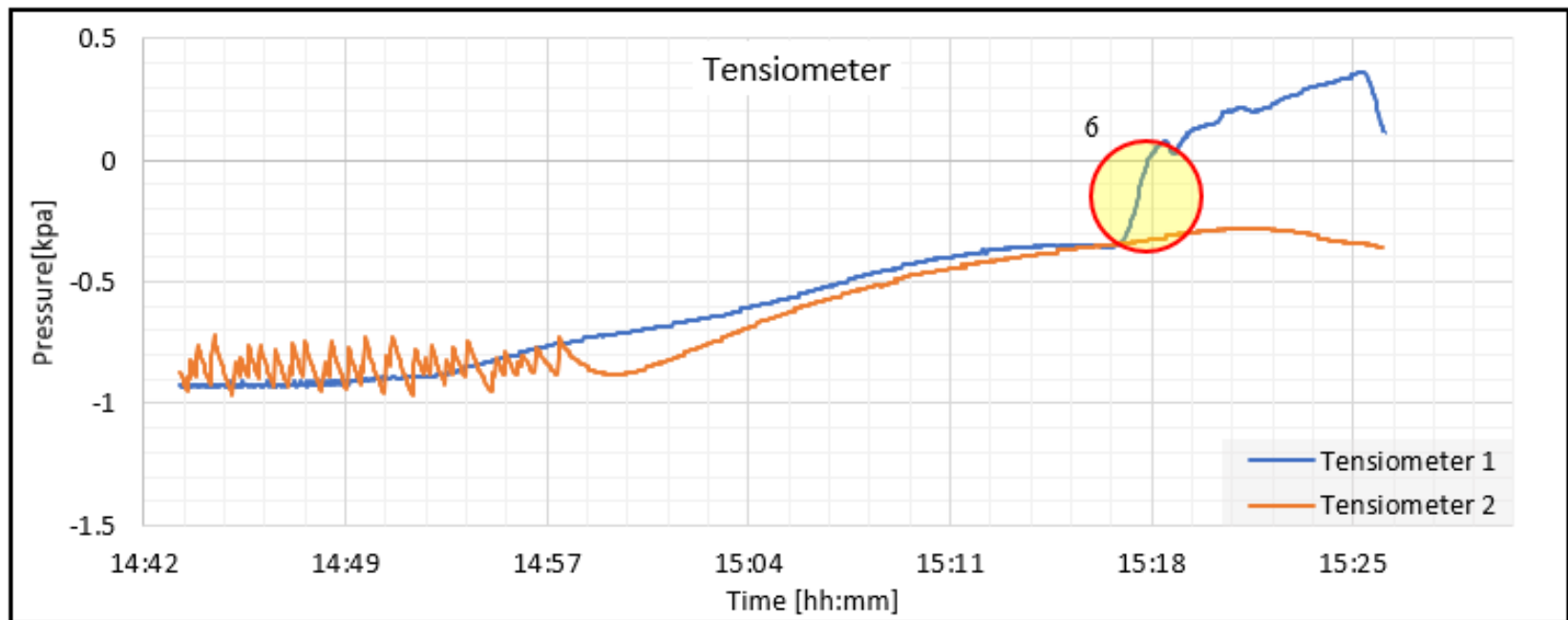


Anomaly 4, Geology



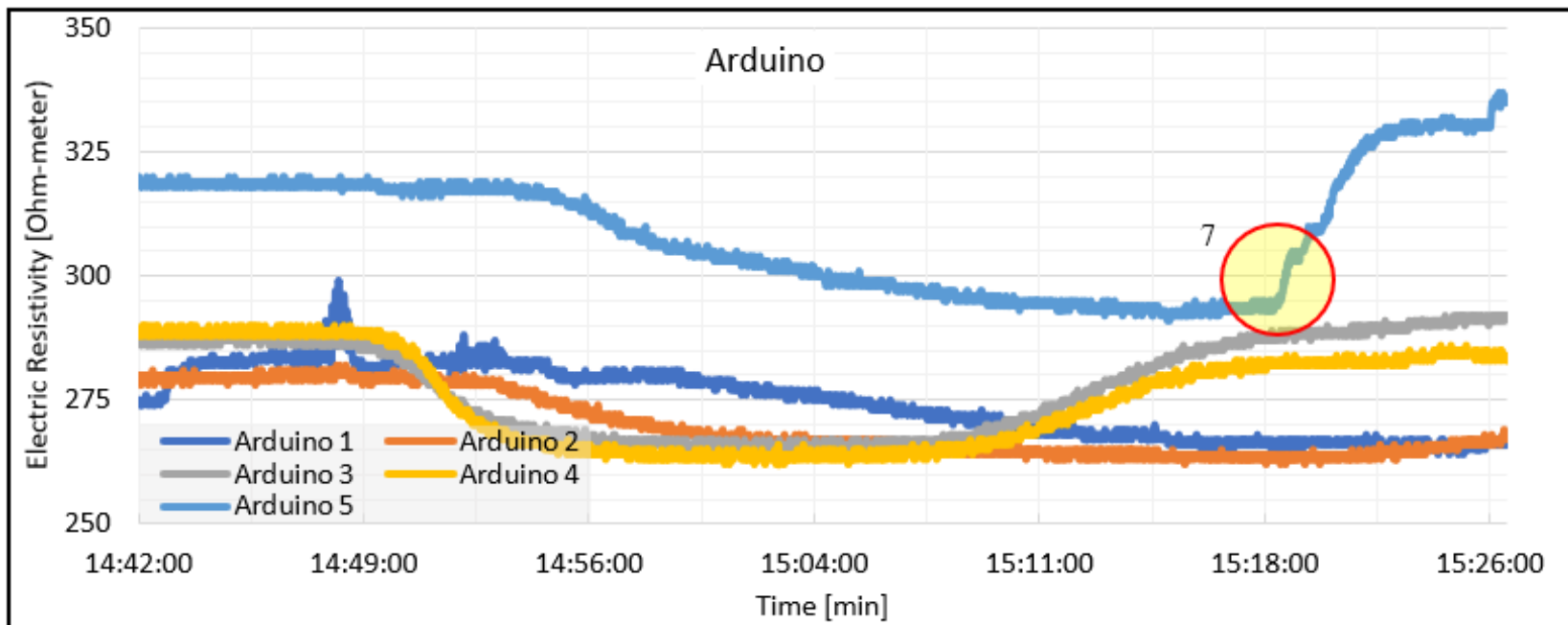


Anomaly 6, Geology





Anomaly 7, Geology

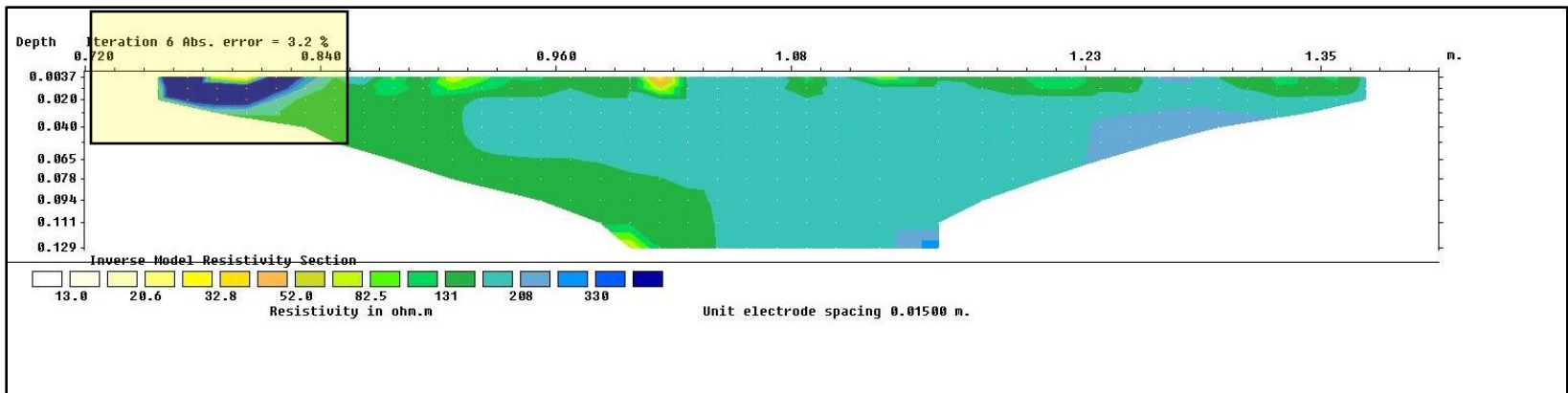
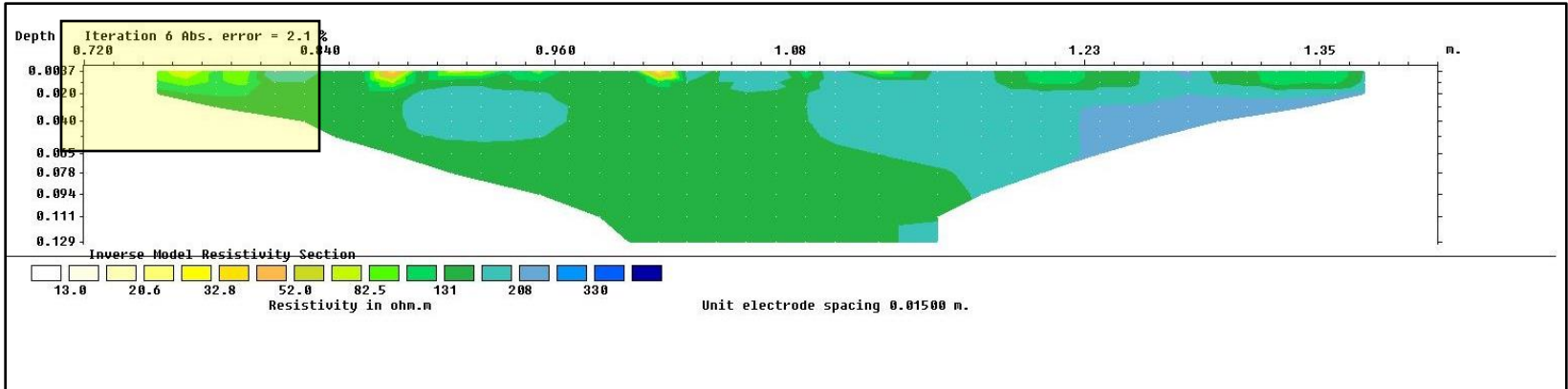


Anomaly 4, 6, & 7, Photogrammetry





Anomaly 4, 6, & 7, Geophysics



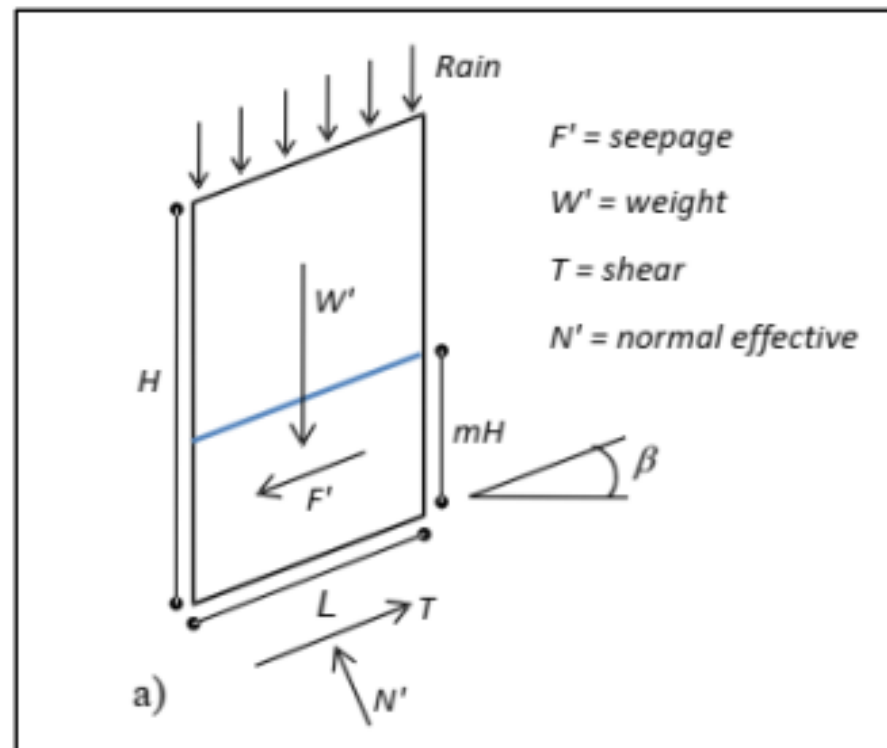


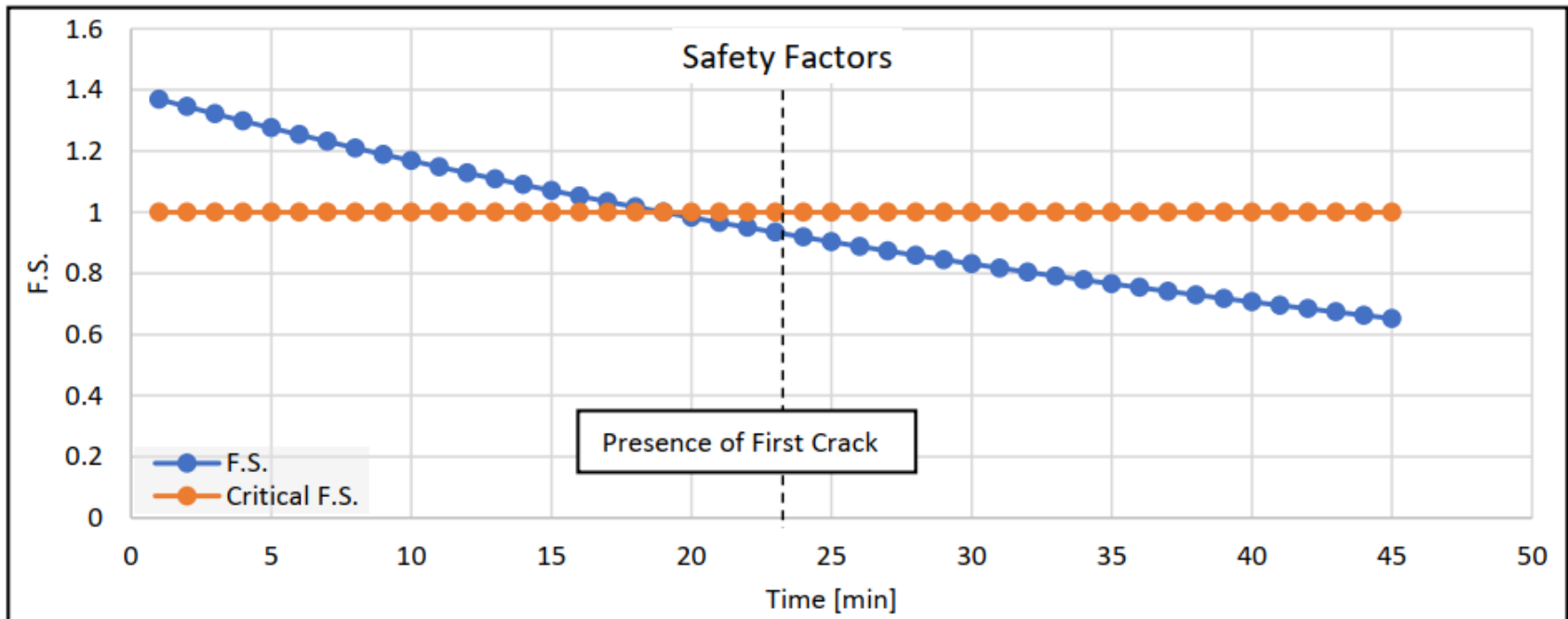
- Statistical Models
- Physical Models
- SLIP model
 - Limit Equilibrium Method
 - 2 sub-layers

$$F_s = \frac{T_s}{T_d}$$

$$T_s = N' + \tan\varphi' + C'$$

$$T_d = W' \sin\beta + F'$$

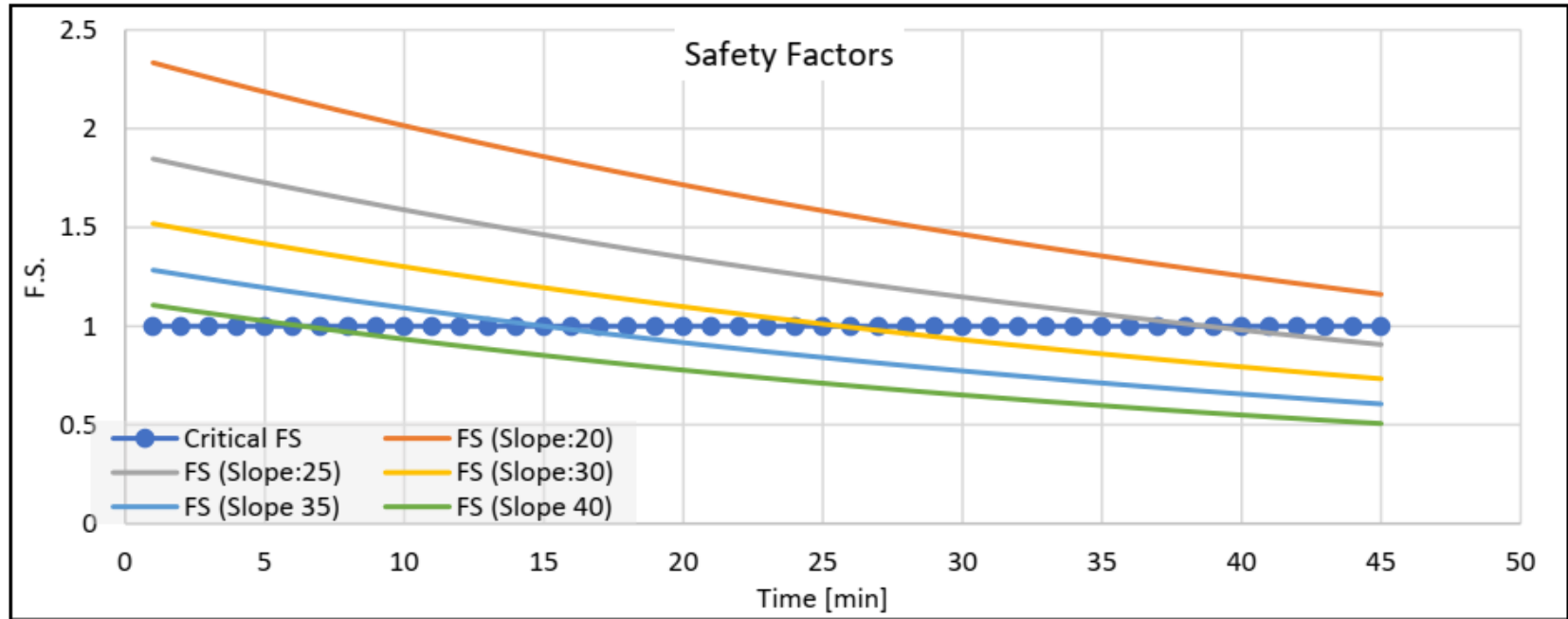




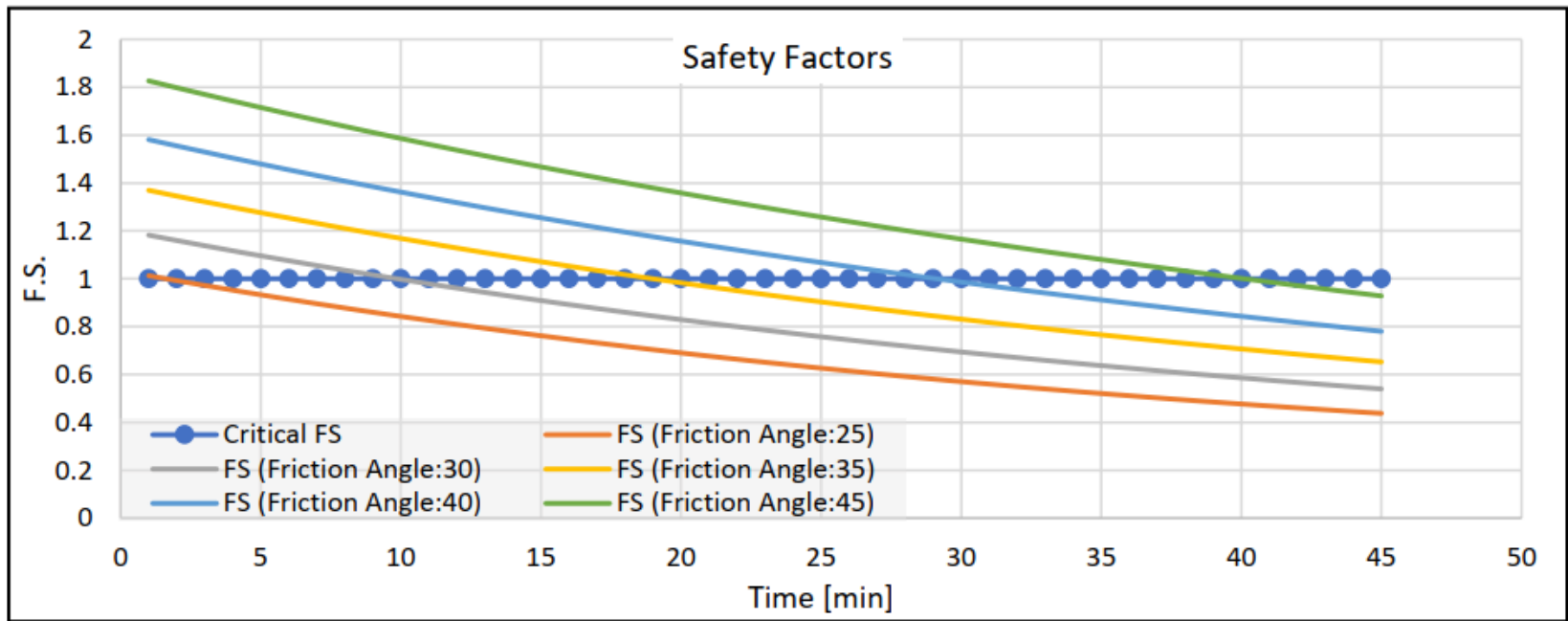
- **4-minute Inaccuracy due to Limitations**
 - Neglecting the run-off water
 - Considering Uniform Layer



SLIP Model: Sensitivity to Slope

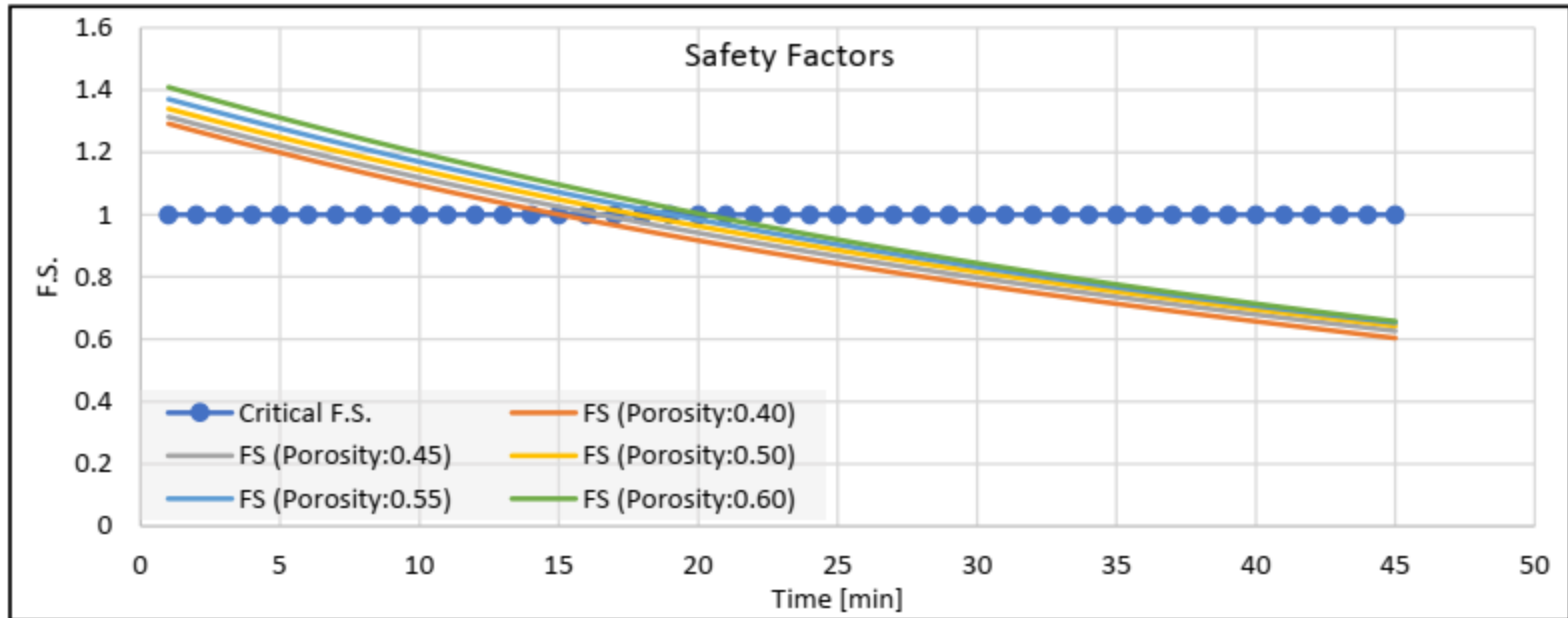


SLIP Model: Sensitivity to Friction Angle

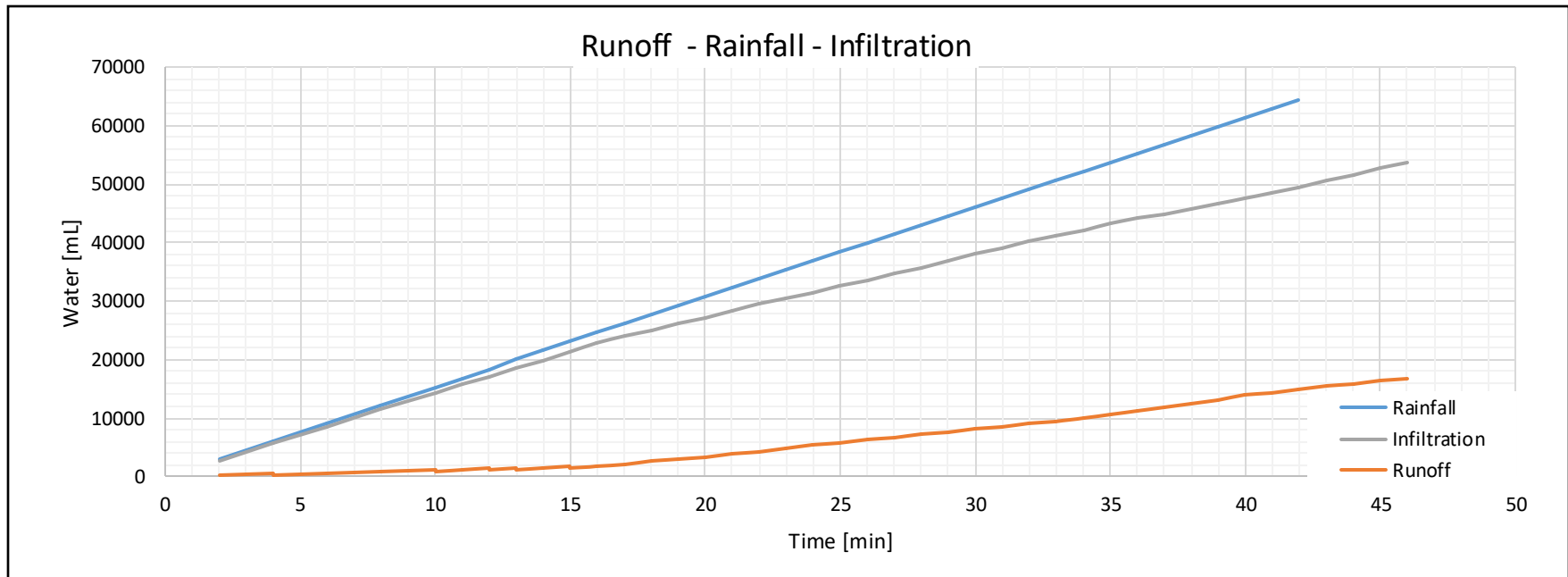




SLIP Model: Sensitivity to Porosity



- Sensitivity Analysis for Different Amounts of Run-off
 - 75% Infiltration, 25% Run-off
 - Confirmed with Collected Run-off Water



- **Default Limit Equilibrium Equation**

$$W' = \cos\beta * H * \Delta s * \gamma_w [m(n - 1) + \cos(1 - n) + nS_r(1 - m)]$$

$$N' = \cos^2\beta * H * \Delta s * \gamma_w [m(n - 1) + \cos(1 - n) + nS_r(1 - m)]$$

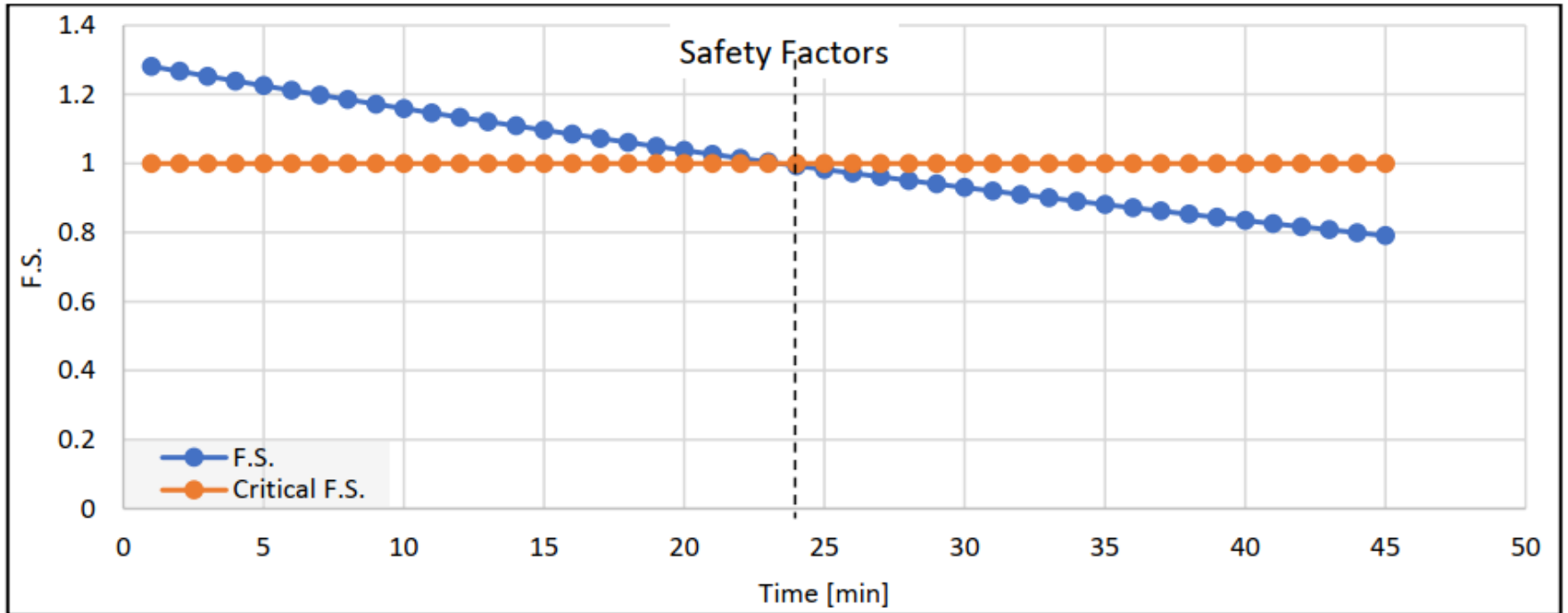
- **Modified Limit Equilibrium Equation**

$$W' = (\cos\beta * 0.5 * H * \Delta s * \gamma_w [m(n - 1) + \cos(1 - n) + nS_r(1 - n)]) + (\cos\beta * 0.5 * H * \Delta s * \gamma_w [m(0.85 * n - 1) + \cos(1 - 0.85 * n) + 0.85 * nS_r(1 - m)])$$

$$N' = (\cos^2\beta * 0.5 * H * \Delta s * \gamma_w [m(n - 1) + \cos(1 - n) + nS_r(1 - n)]) + (\cos^2\beta * 0.5 * H * \Delta s * \gamma_w [m(0.85 * n - 1) + \cos(1 - 0.85 * n) + 0.85 * nS_r(1 - m)])$$



Modified SLIP Model Results





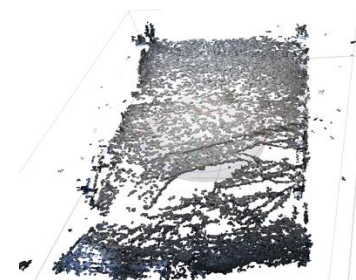
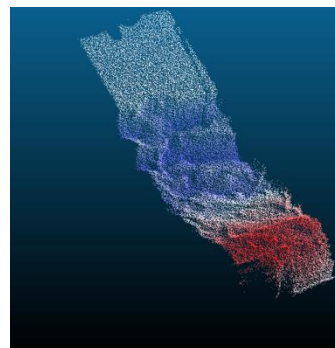
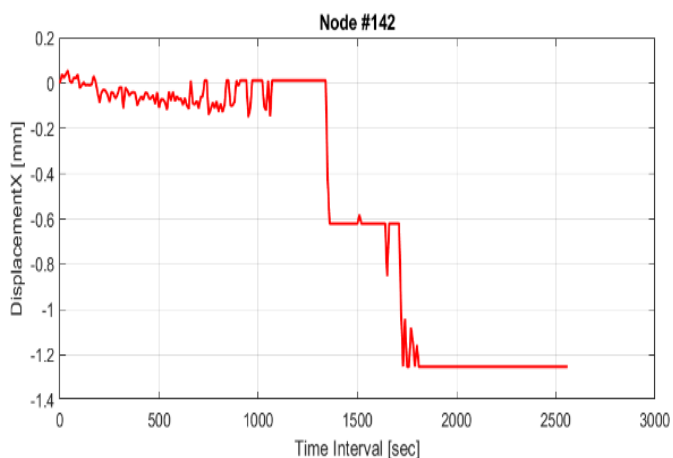
Suggestions for Better Experiment



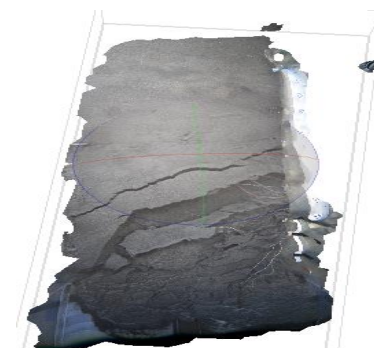


Photogrammetry

- Definition of Photogrammetry
- Uses of Photogrammetry in Landslide Experiment
 - Spatial data of the fractures



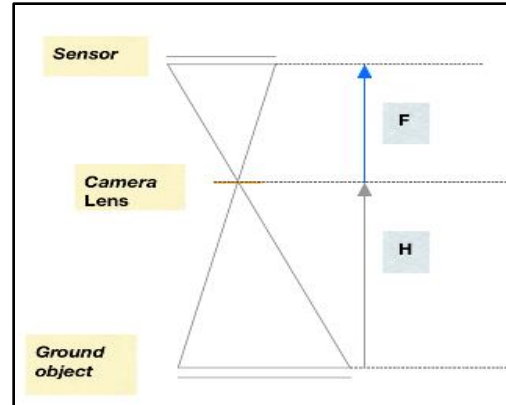
- Point cloud of the Landslide



- **2-D Digital Image Correlation**
- **Structure for Motion**
- **Terrestrial Laser Scanner data**

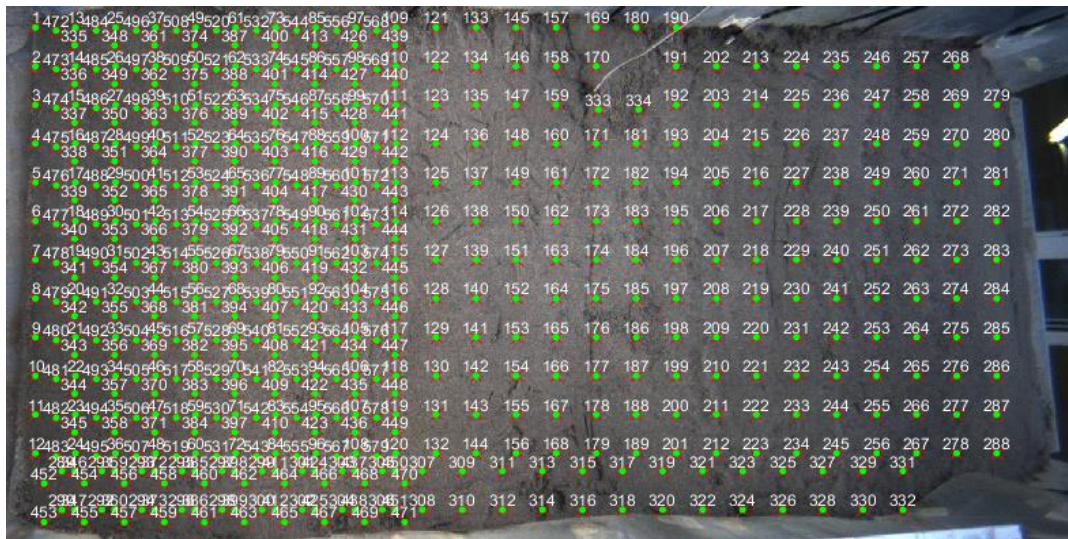
- 2D Digital Image Correlation

GSD PARAMETERS



$$GSD = 0.6325 \text{ [MM/PIX]}$$

GRID SELECTION

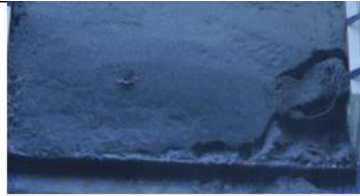




Experiment 1 Results



Front Camera, Fracture 1



Front Camera, Fracture 2



Front Camera, Fracture 3



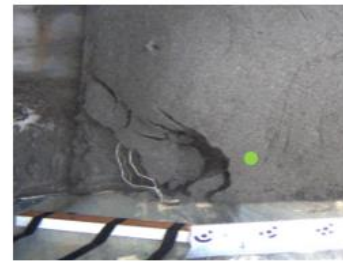
Front Camera, Fracture 4,5,6



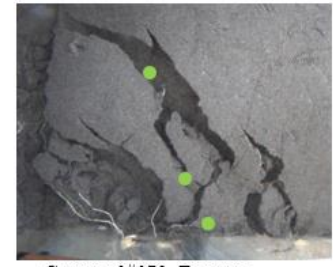
Camera 1#103, Fracture 1



Camera 1#148, Fracture 2



Camera 1#164, Fracture 3

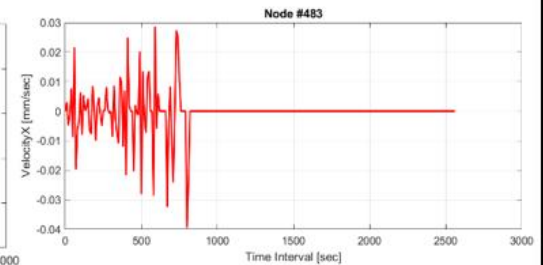
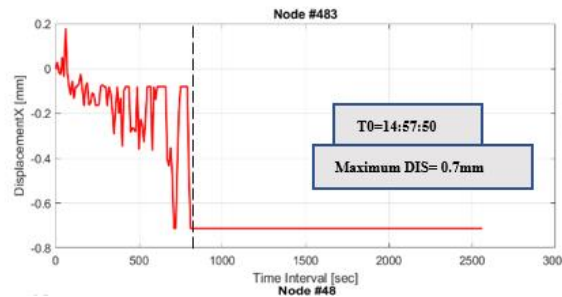


Camera 1#173, Fracture 4,5,6

Fracture 1

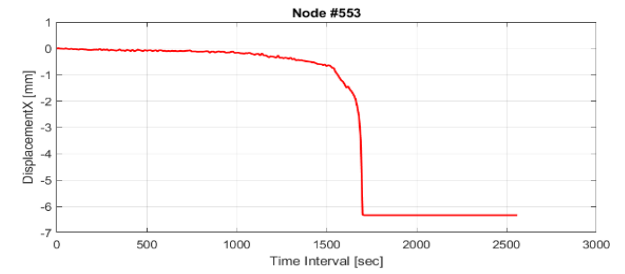
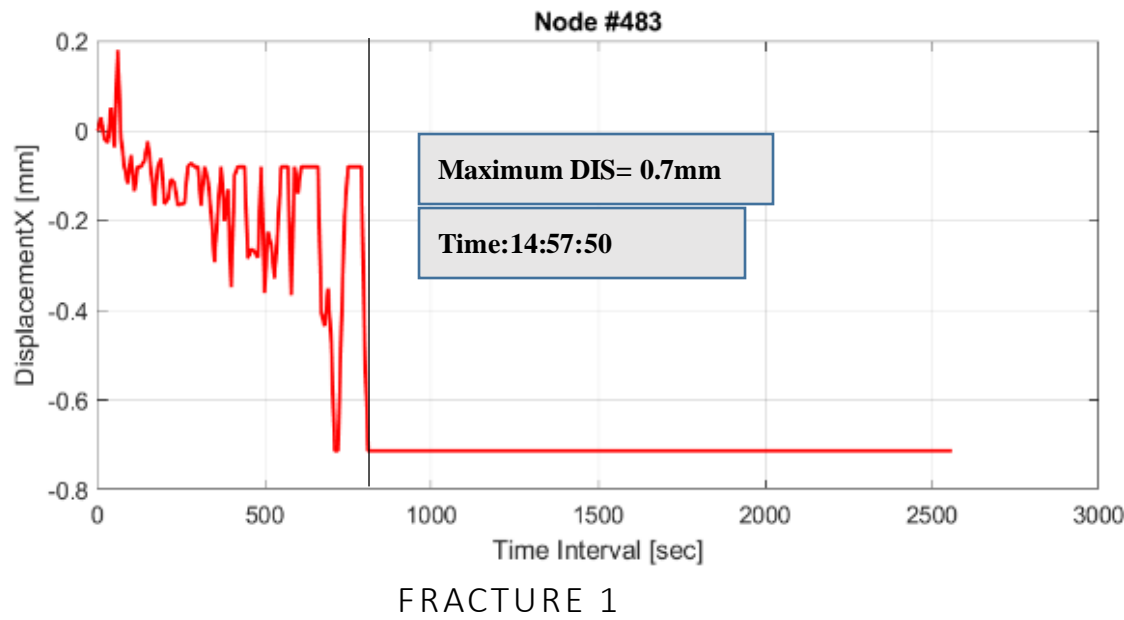
Coordinates of node 483 (X=6.92, Y=9.84)

-Progressive Failure

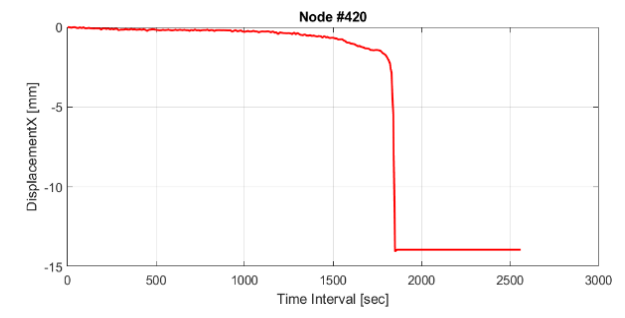




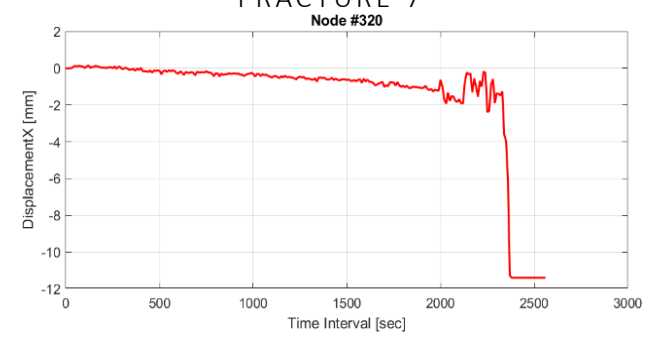
Displacement Results



FRACTURE 4



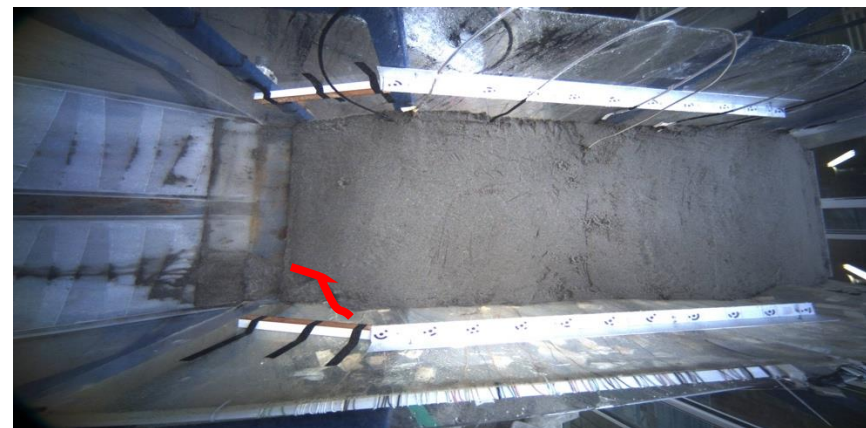
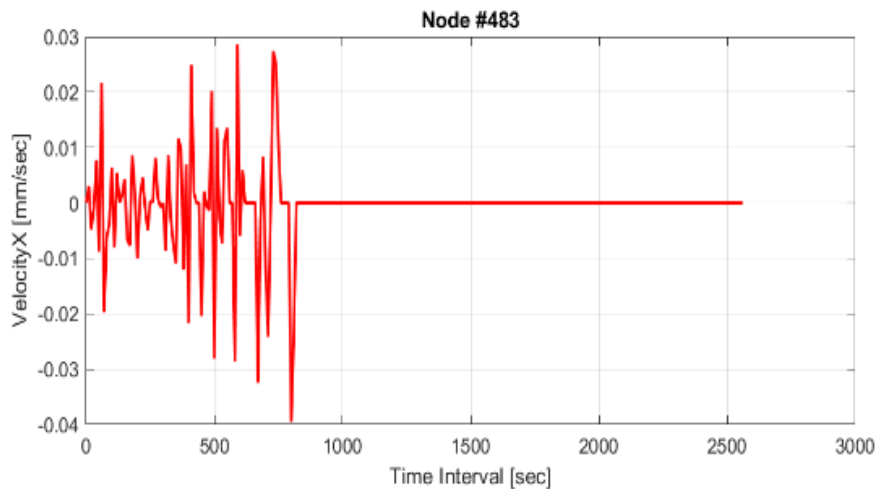
FRACTURE 7



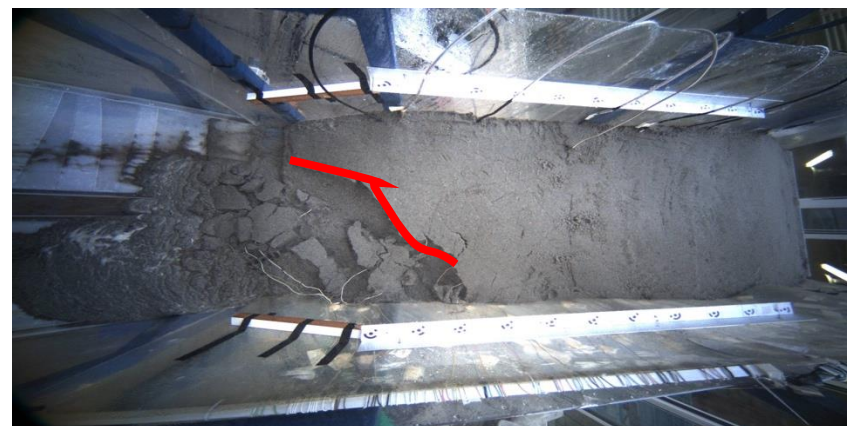
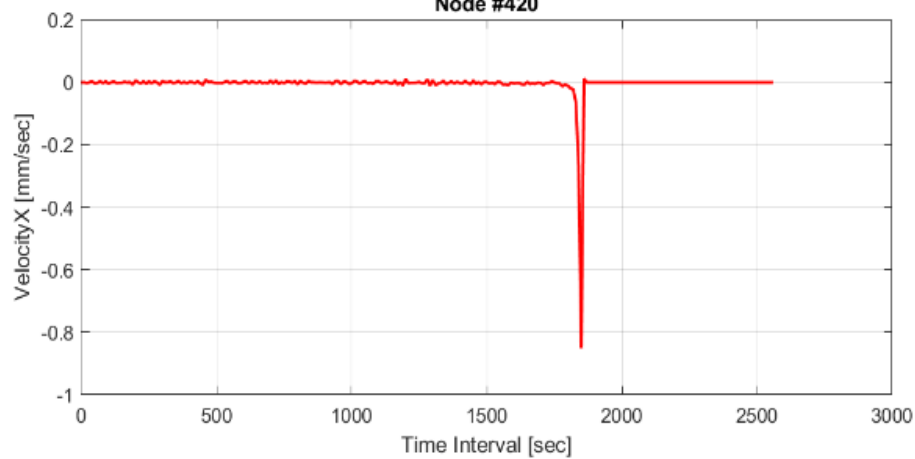
FRACTURE 13



Velocity Results



FRACTURE 1
Node #420

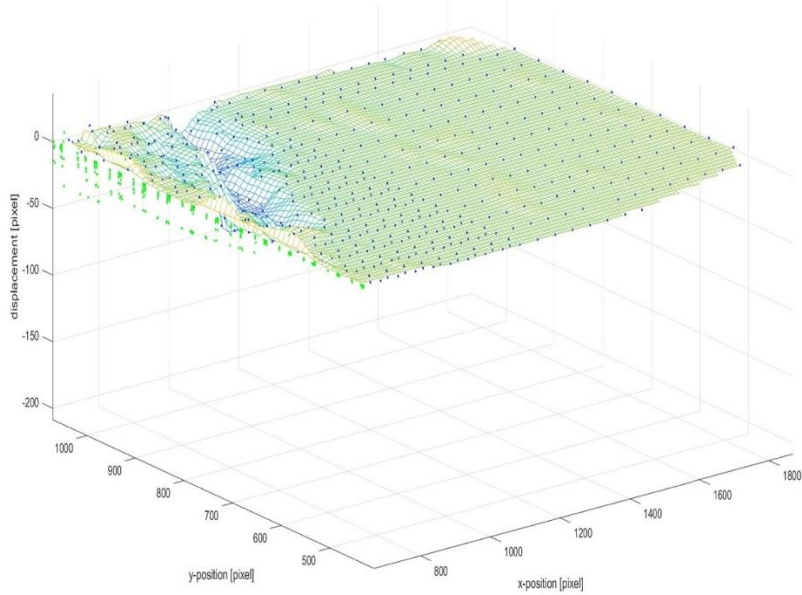


FRACTURE 7

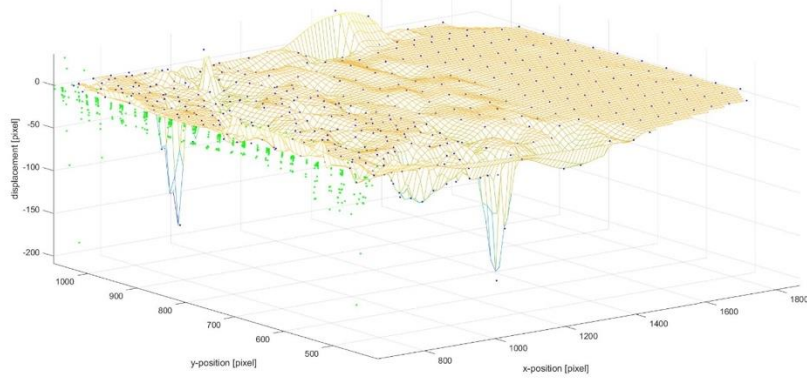


3D Image Displacement

Displacement versus x-y-position (Current image #: 173)



Displacement versus x-y-position (Current image #: 242)





Failure Phases

TIME DIFFERENCE
BETWEEN
FRACTURE 1 AND
2

Time:14:57:50

Time:15:08:50

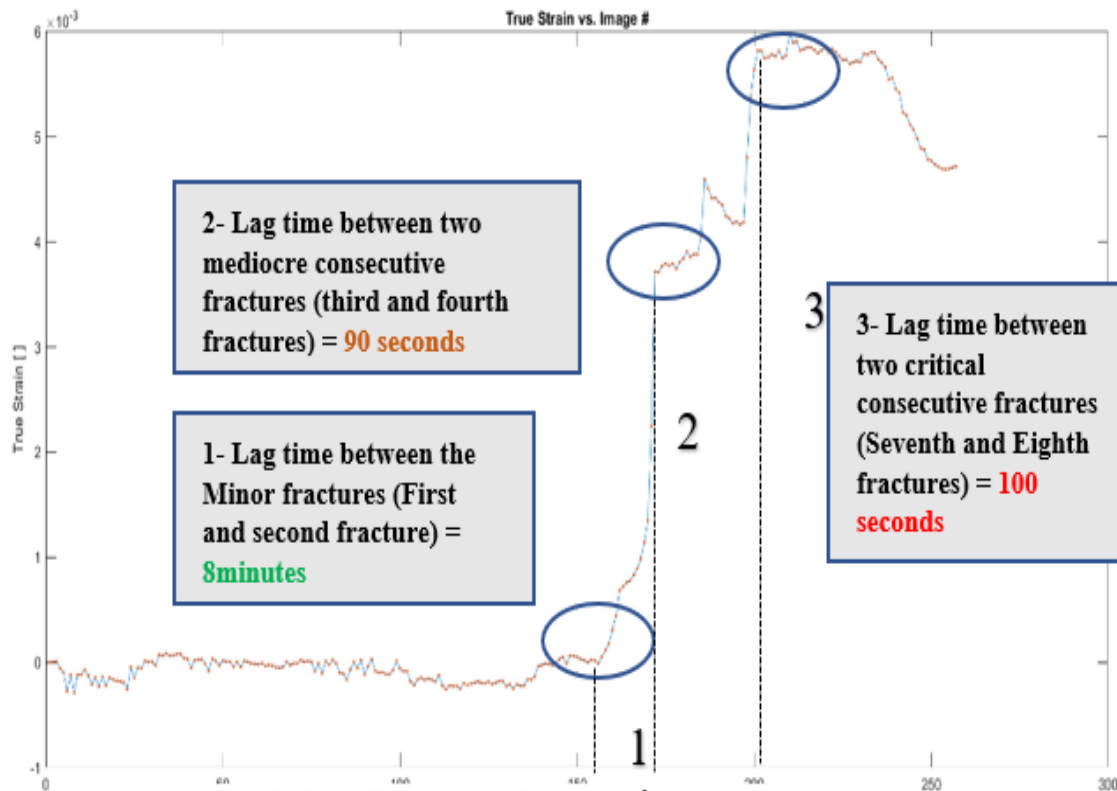
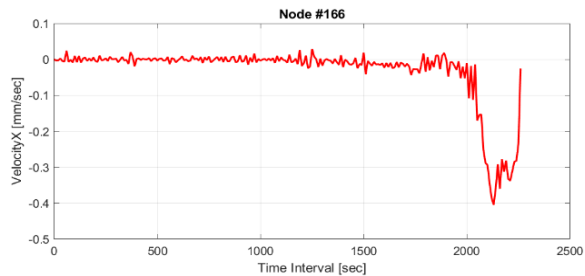
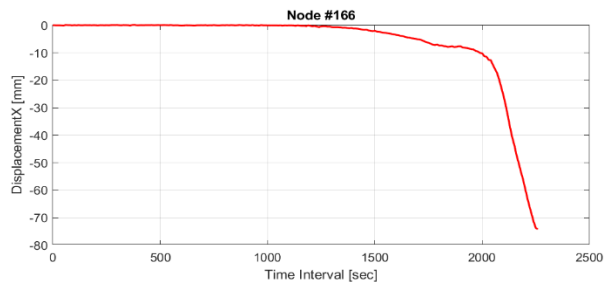
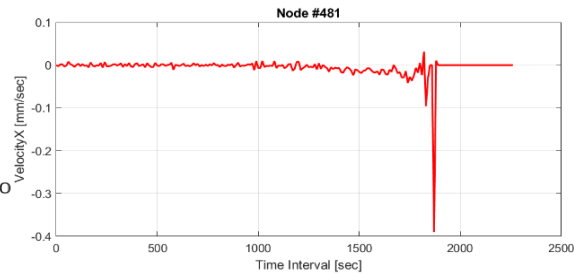
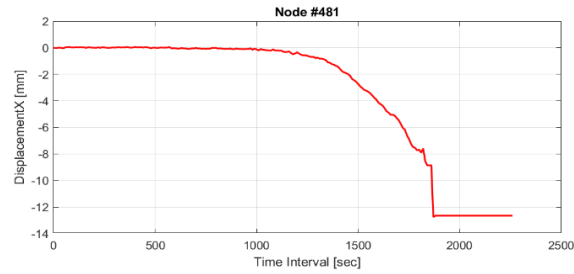


Fig.49. Exp1, True strain vs Image number

Fracture Number	Electrode Number	VWC (Geophysics)	VWC (Geology)
Fracture 2	1	0.36	0.2975
Fracture 3	3	0.45	0.3201
Fracture 4	24	0.31	0.3284

Experiment 0 vs Experiment 1



- **Experiment 0**

1- GRADUAL INCREASE
FOLLOWED BY SUDDEN
FAILURE

2- SMALL TIME DIFFERENCE
BETWEEN DIFFERENT
FRACTURES(MORE
AGGRESSIVE)



- 3D IMAGE GENERATION

- Cameras Calibration



Camera type: Frame

Pixel size (mm): 0.00345 x 0.00345

Focal length (mm): 3

Rolling shutter compensation: Disabled

Film camera with fiducial marks

Initial Adjusted Bands GPS/INS Offset

Type: Precalibrated

cx: -24.0753

f: 656.151913 cy: -22.5855

k1: -0.00323497 p1: -0.000127756

k2: 0.00602171 p2: 0.000509693

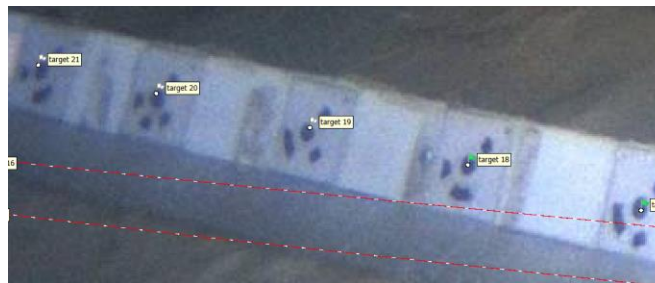
k3: -0.00175929 b1: 0

k4: 0 b2: 0

Fixed parameters: All Select...

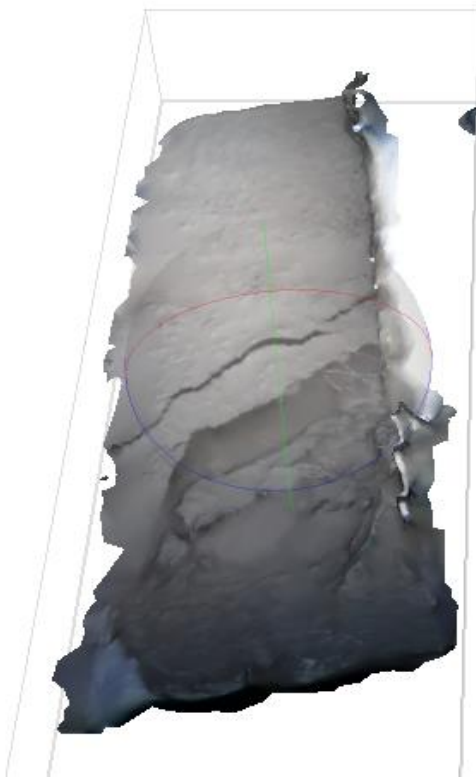
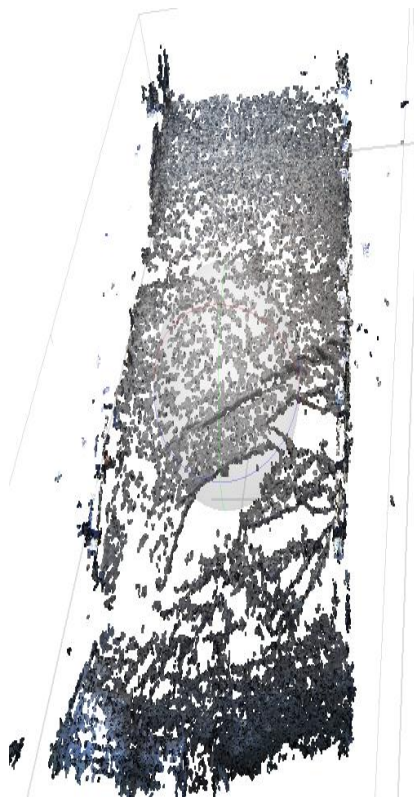
Image-variant parameters: None Select...

- Photo Alignments





3D Image for Fracture 8





Orthophotos for Fracture 7

- Orthophotos

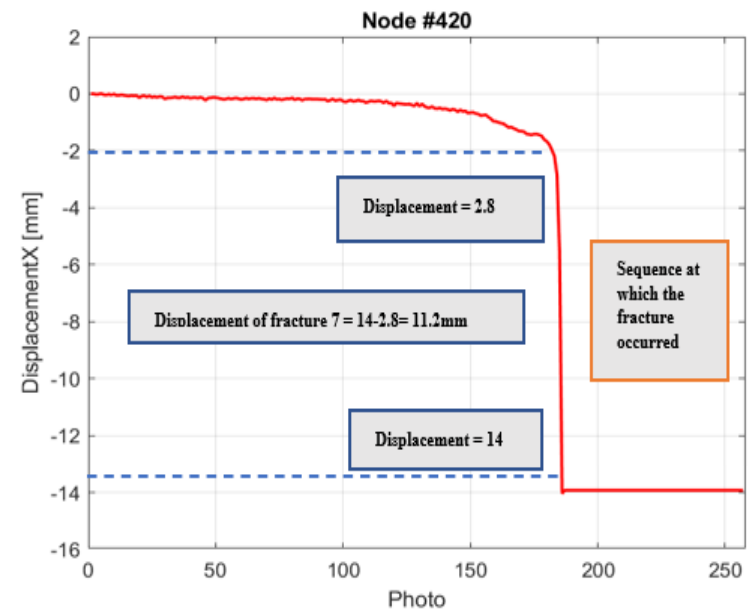
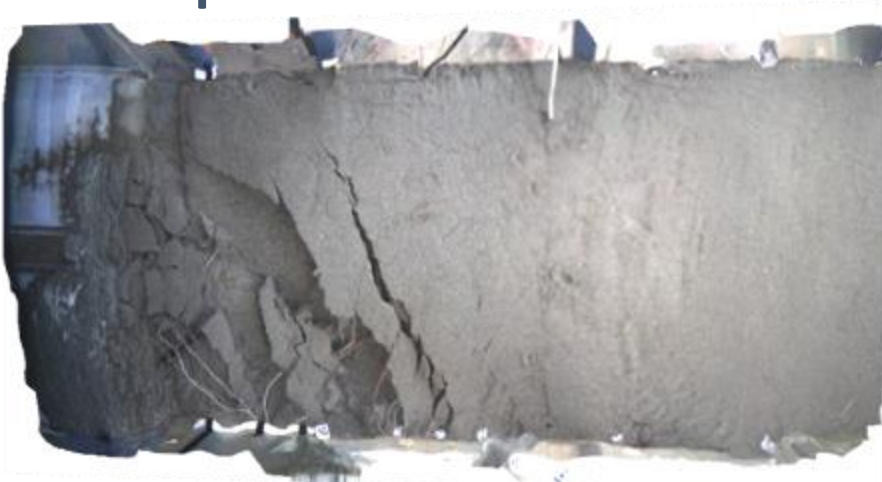
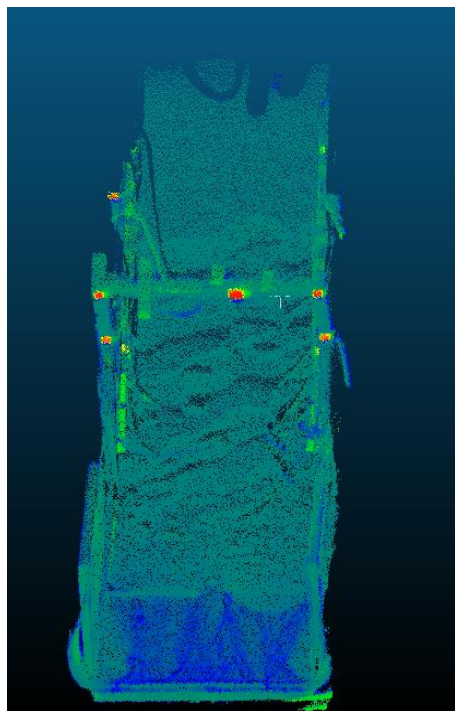


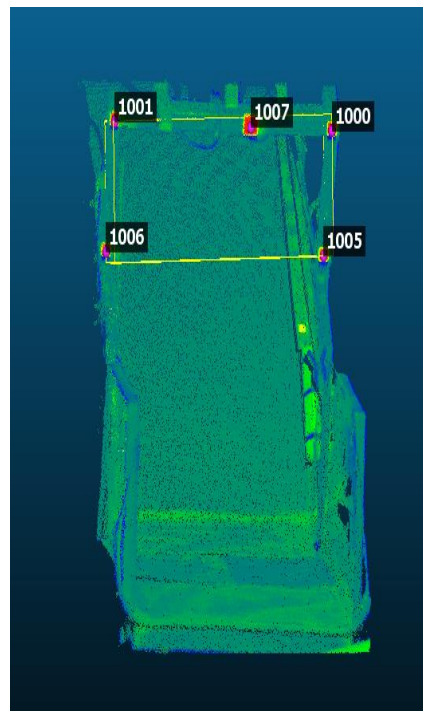
Fig.73. Displacement as measured for fracture 7 from 2D DIC



- **Filtering**

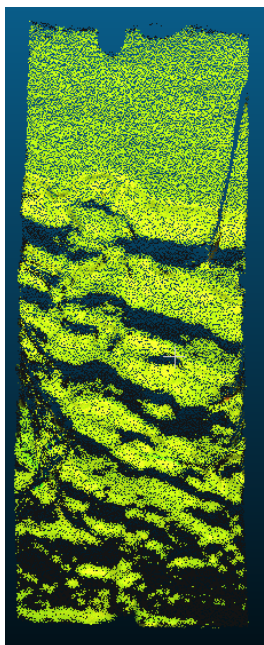


- **Geo-referencing**

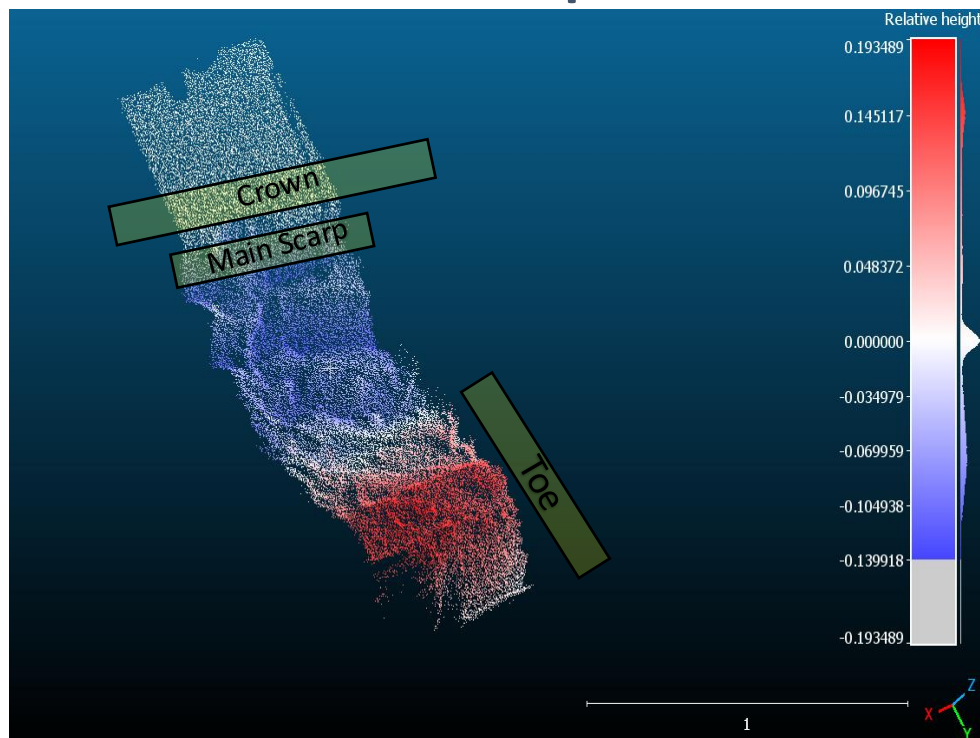




- Normal Vectors



- Volume Computation



step: 0.004000

size: 195 x 513 (100,035 cells)

projection dir.: Z

cell height: average height

Results

At least one of the cloud is sparse!
You should fill the empty cells...

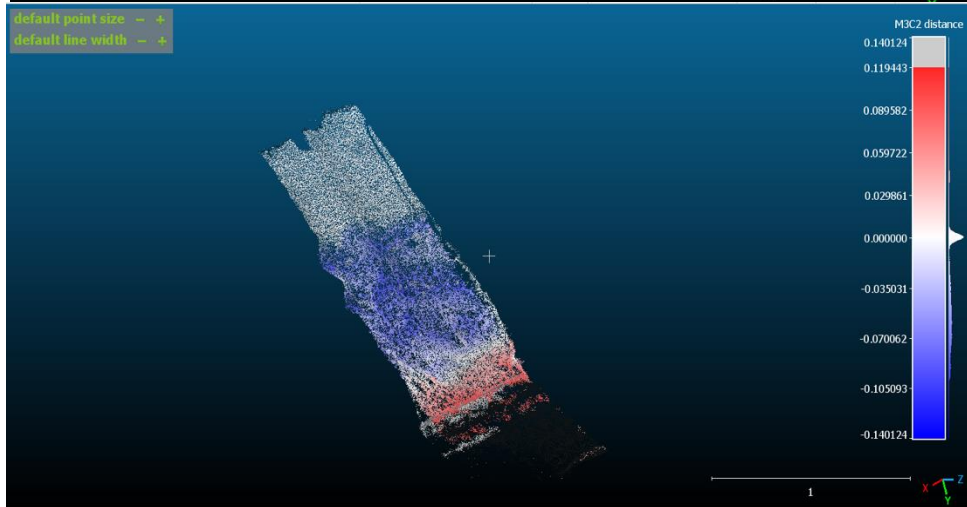
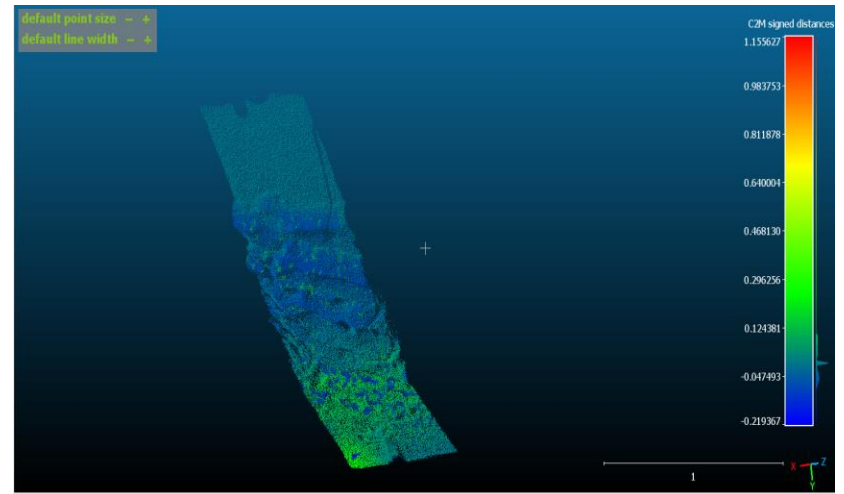
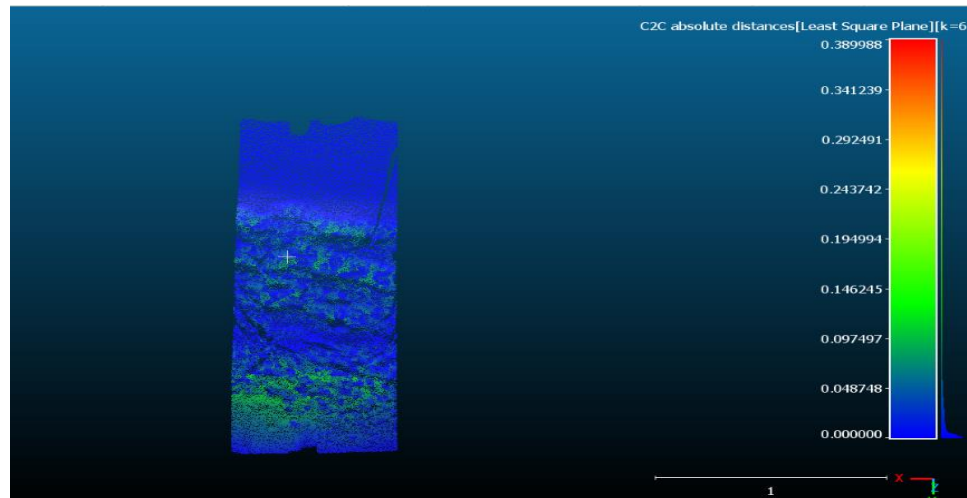
Volume: 0.005
Surface: 0.748

Added volume: (+)0.026
Removed volume: (-)0.021

Matching cells: 64.3%
Non-matching cells:
ground = 16.9%
cell = 18.7%

Num. precision: 3

Distance Difference Computation

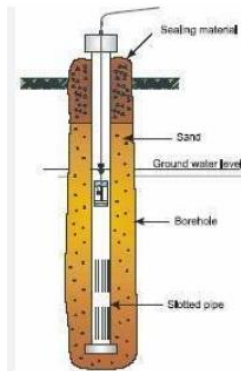




Early Warning System



Fracture Number	Electrode Number	VWC (Geophysics)	VWC (Geology)
Fracture 2	1	0.36	0.2975
Fracture 3	3	0.45	0.3201
Fracture 4	24	0.31	0.3284



VWV: 0.18

4 minutes in between

Lead Time: 5 Minutes before minor fractures

Lead Time: 9 Minutes before the more mediocre fractures

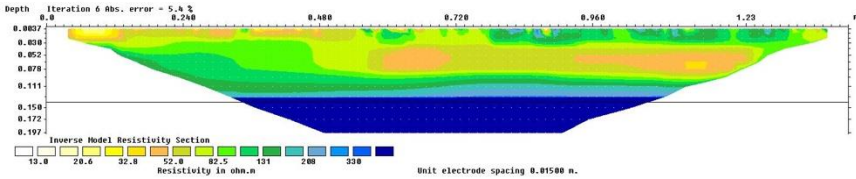


- DC resistivity method
- Electrical Resistivity Tomography(ERT)
- Wenner array

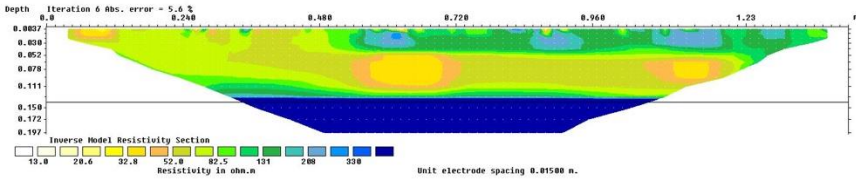


Resistivity Model in Experiment 1

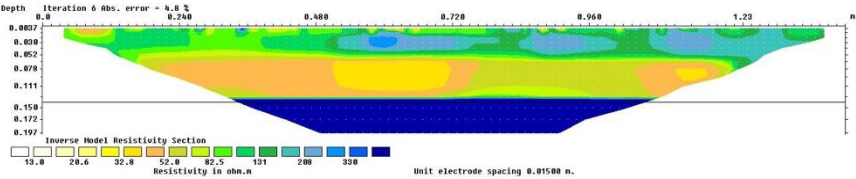
T2 at 6 min



T3 at 11 min



T4 at 17 min



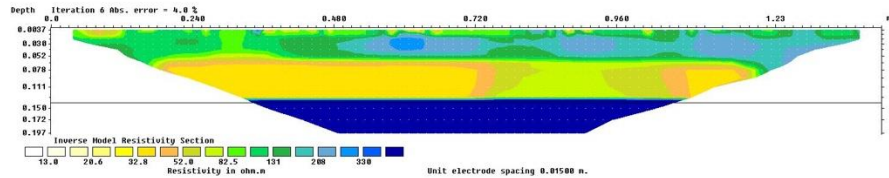
Phenomenon:

1. whole slope resistivity of soil is **decreasing**, especially the toe of the slope
2. the decrease in soil resistivity **wasn't in a smooth trend.**

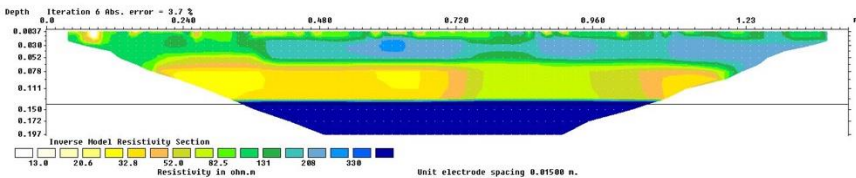


Analysis for Resistivity Model of Software

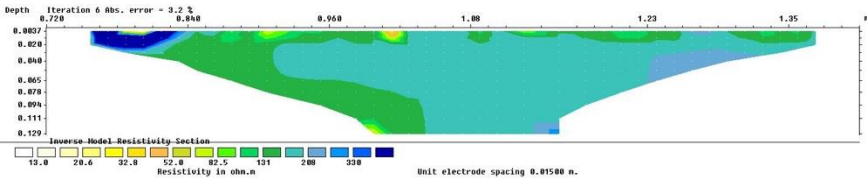
T5 at 22 min



T6 at 27 min



T10 at 39 min



Phenomenon :

1.the soil resistivity began to **increase in the crown part** meanwhile the **toe part kept decreasing**.

2.in the increasing part, the resistivity is still **smaller than $200\Omega\text{m}$** , we cannot consider it as the real crack but only the zones of different saturation.



Mathematical Resistivity Model

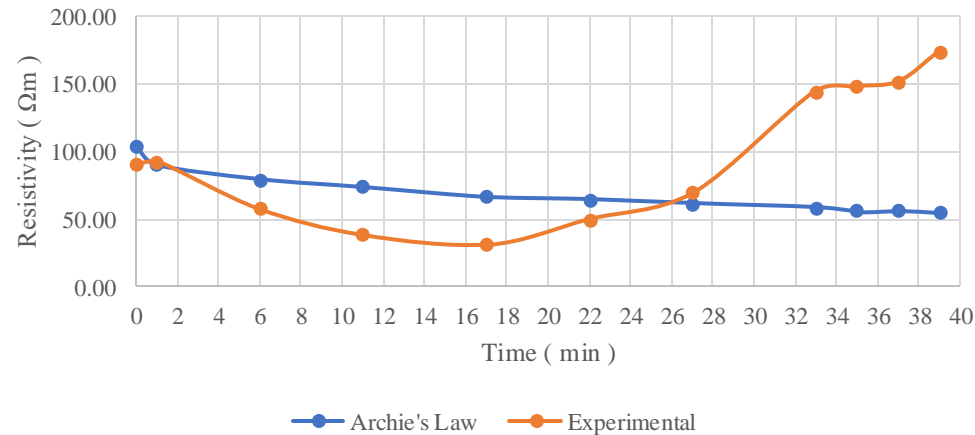
-Archie's Law

$$\rho = F S_w^{-n} \rho_w$$

we can get the calibrated values for a, m, n:

a	m	n
0.5	1.3	1.2

Resistivity comparison from T0-T10



relative difference is quite high.

Why?

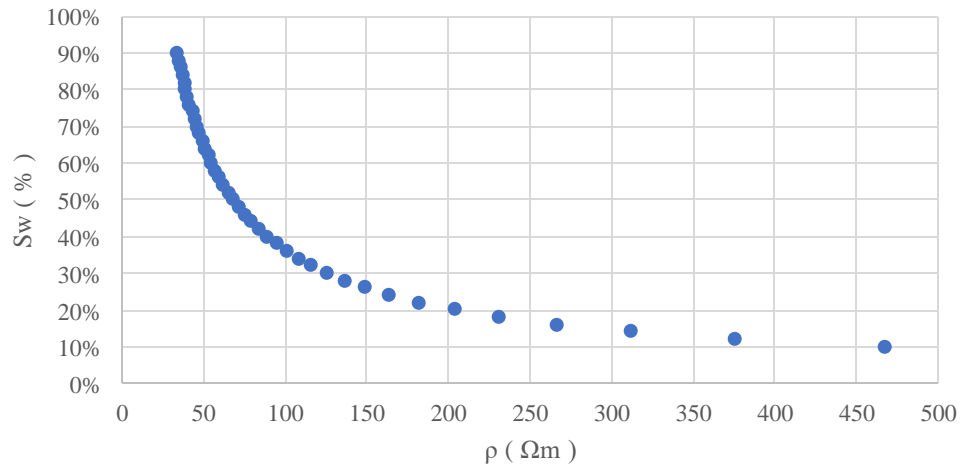
1. not uniformly compacted soil
2. the flaws that happened in this experiment is that the crack position is too far from electrodes.
3. during the experiment, the time when we tested the resistivity from the electrodes is not completely accurate compared with the time measuring porosity and volumetric water content for geology parts. Because the parts is operated by different people.

Water Saturation vs. Resistivity

As a result, we developed the following equation to obtain the water saturation of the landslide body from inverted resistivity data :

$$S_w = \left(\frac{29.43}{\rho}\right)^{0.83}$$

Water saturation VS Resistivity



dry sand to the water saturation of about 30%-40%, the resistivity values **rapidly decrease** from 500 Ωm to 100 Ωm approximately .

when water saturation beyond about 40%, the resistivity **decreases softly and gently**.

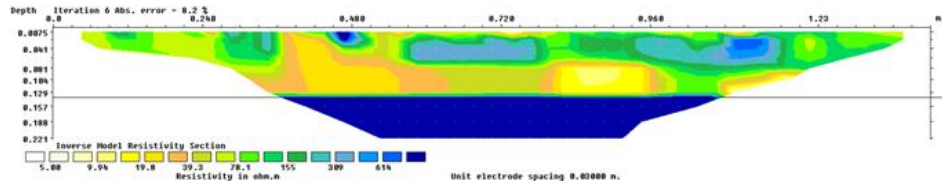
Until the soil is almost **full of water**, the **resistivity is getting closer to zero**.

From the graph we can easily observe that **resistivity is always decrease while the water saturation increasing**, but the relationship between the two isn't linear.

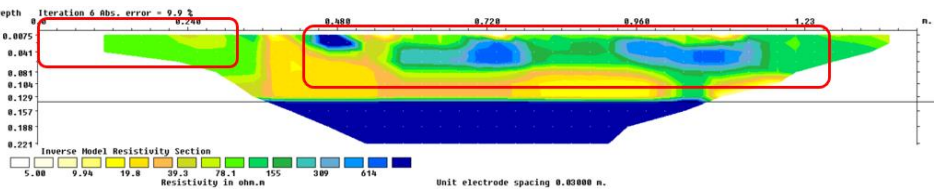


Resistivity Model in Experiment 0

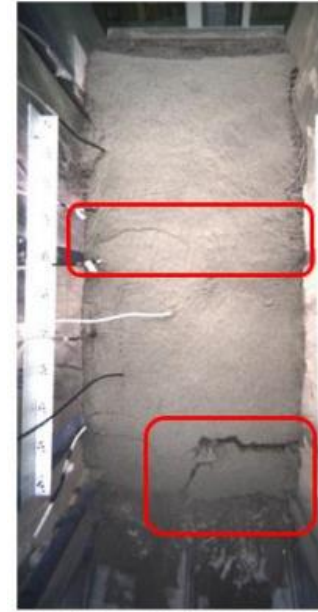
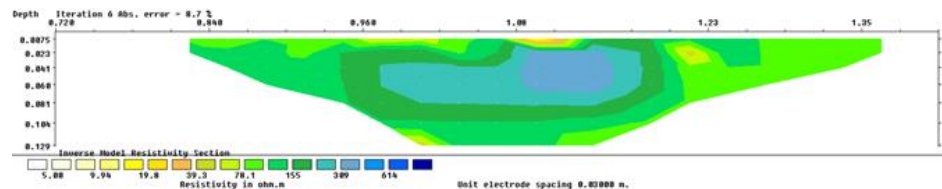
T5 at 28 min



T6 at 33 min



T7 at 38 min



T6

Phenomenon :

other parts resistivity is decreasing ,while in T6 we can easily find that resistivity is increasing and over $200\Omega\text{m}$

To conclude, it's obvious that the results obtained from geology, photogrammetry and geophysics were supplementary to each other.



Field Trip Geology



Figure 3. Family 2



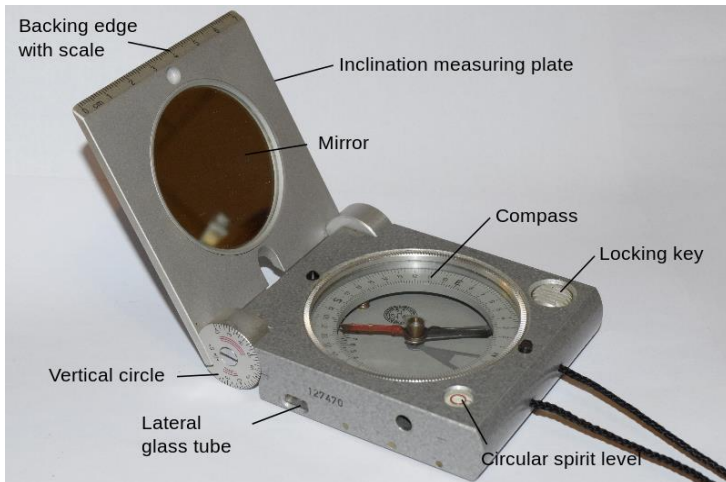
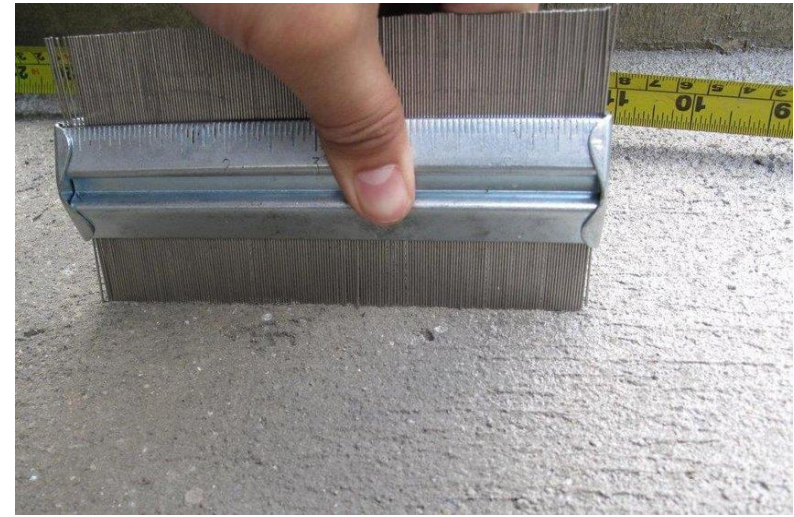
Figure 2. Family 1



Figure 4. Family 3



Instruments



- Joint Orientation and Data Processing
- Joint Spacing
- Intercept
- Water presence
- Infill
- Persistence
- Weathering
- Estimation of the Joint Roughness Coefficient (JRC).
- Joint Compressive Strength (JCS)
- Rock Quality Designation (RQD)
- Aperture

Aperture is the perpendicular distance separating the adjacent rock walls of an open discontinuity, in which the intervening space is air or water filled. A classification of aperture is shown in table [5.]. Moreover, the measure of aperture between discontinuities of all 3 families demonstrate very wide or open as it shown in the table [6.].

Table 1. Classification of the separation of joints (ISRM, 1978).

Table 2. Aperture of joint sets

Term	Separation
Very tight	<0.1 mm
Tight	0.1 – 0.5 mm
Moderately	0.5 – 2.5 mm
Open	2.5 – 10 mm
Very open	10 – 25 mm
Wide	>10 mm
Very wide	1 – 10 cm
Extremely wide	10 – 100 cm
Cavernous	>1 m

Joint set	Aperture	Classification
K1	<1 mm	Very tight
K2	<1 mm	Very tight
K3	<1 mm	Very tight



Joint Families

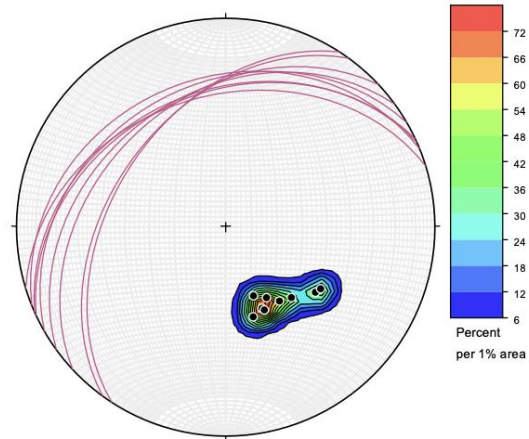


Figure 9. Family 1

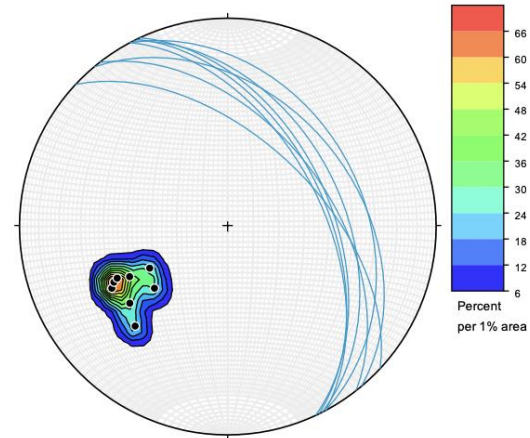


Figure 10. Family 2

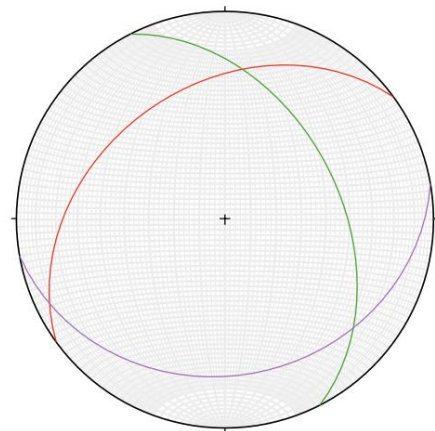


Figure 12. Family Mean Vector

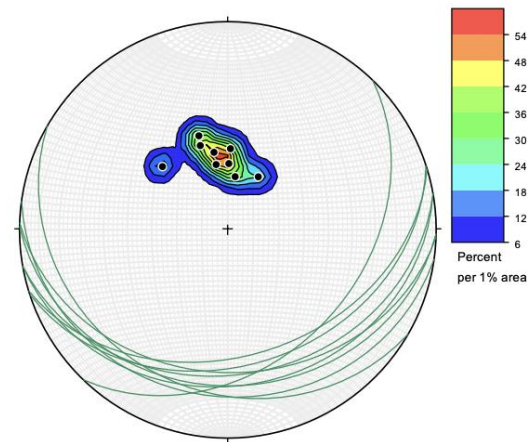


Figure 11. Family 3



Kinematic analysis: Circular and Planar Failure

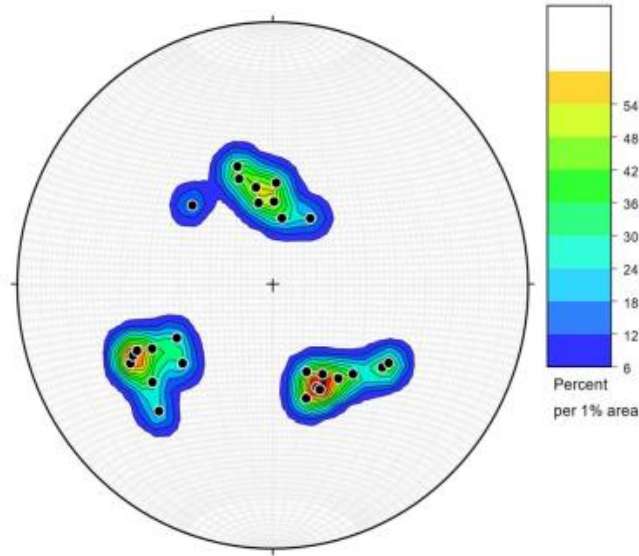


Figure 18. Poles of All Joint Sets

K1	36	324
K2	44	63
K3	26	170
slope	63	134
Friction angle	35	

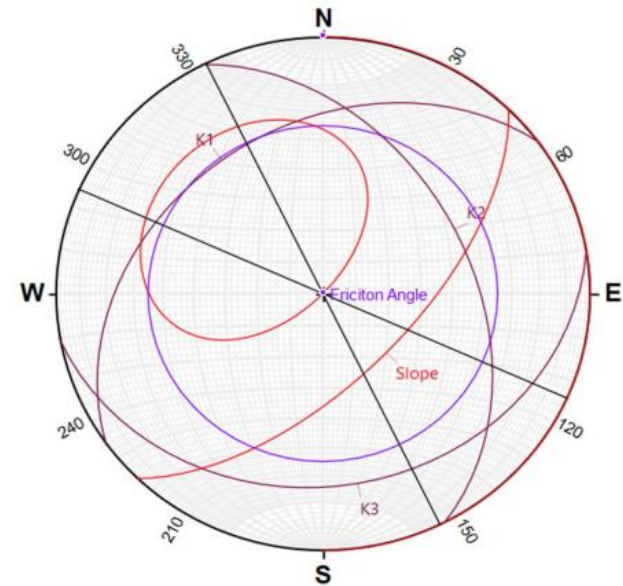


Figure 19. Mean orientation of joint sets, slope and friction angle in Stereonet

Planar failure	K1	K2	K3
Condition 1	yes	yes	yes
Condition 2	no	no	no
Condition 3	yes	yes	no
Condition 4	yes	no	no



Kinematic analysis: wedge failure and Toppling

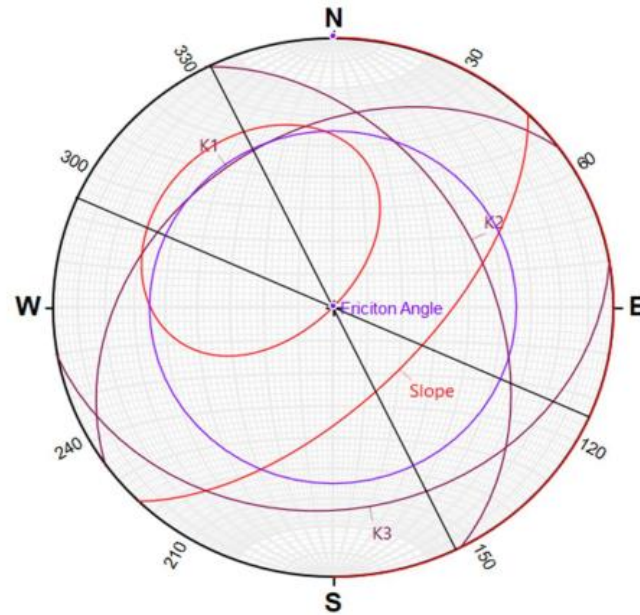


Figure 19. Mean orientation of joint sets, slope and friction angle in Stereonet

Wedge failure	K1 & K2	K2 & K3	K1 & K3
Intersection line/ dip	29	21	8
Condition 1	yes	yes	yes
Condition 2	no	no	no
Condition 3	no	no	no

Toppling	K1	K2	K3
Condition 1	no	no	no
Condition 2	no	yes	no



Block volume estimation

Persistence Reduction:

$$V_b = \frac{S_1 \times S_2 \times S_3}{\sin \gamma_1 \times \sin \gamma_2 \times \sin \gamma_3 \sqrt{p_1 \times p_2 \times p_3}}$$

$$V_b \text{ average} = \frac{S_1 \times S_2 \times S_3}{\sin \gamma_1 \times \sin \gamma_2 \times \sin \gamma_3} \longrightarrow$$

$$V_b = 756,898 \text{ cm}^3 \sim 0.757 \text{ dm}^3$$

$$V_b = 5.236,36 \text{ cm}^3 \sim 0.0052 \text{ dm}^3$$

$$V_b \text{ Minimum} = \frac{S_1 \times S_2 \times S_3}{\sin \gamma_1 \times \sin \gamma_2 \times \sin \gamma_3} \longrightarrow$$

$$V_b = 61 \text{ cm}^3 \sim 0.06 \text{ dm}^3$$

$$V_b = 436,36 \text{ cm}^3 \sim 0.436 \text{ dm}^3$$

$$V_b \text{ Maximum} = \frac{S_1 \times S_2 \times S_3}{\sin \gamma_1 \times \sin \gamma_2 \times \sin \gamma_3} \longrightarrow$$

$$V_b = 3932 \text{ cm}^3 \sim 3.932 \text{ dm}^3$$

$$V_b = 27.927 \text{ cm}^3 \sim 0.027 \text{ dm}^3$$

Rock volume categorization	
Very small	$V_b = 10 - 200 \text{ cm}^3$
Small	$V_b = 0.2 - 10 \text{ dm}^3$
Moderate	$V_b = 10 - 200 \text{ dm}^3$
Large	$V_b = 0.2 - 10 \text{ m}^3$
Very large	$V_b > 10 \text{ m}^3$

Table 22. Classification of block volume suggested by Palmstrom (1995)

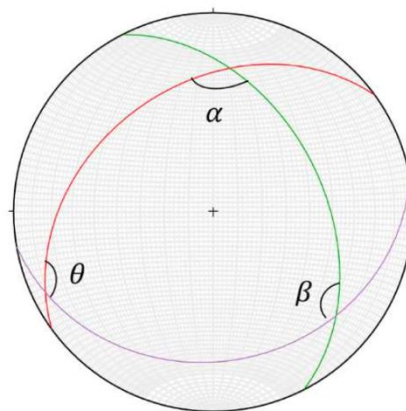


Figure 20. Representation of angles between discontinuities



Safety distance evaluation of rockfall

$$Ra = 9.98 Vb^{0.33}$$

We have considered the safety distance for three block volume:

Size	Volume (cm ³)	Safety distance (Ra) (cm)
Average	759.89	88.9
Minimum	61	38.72
Maximum	3932	153

Table 23. Safety Distance of Blocks

Additionally, we have calculated the X distance (the reachable distance of the block from the position of its detachment) for the three values of the Vb, while considering an assumption of an average height of 1m for the elevation of the rock.

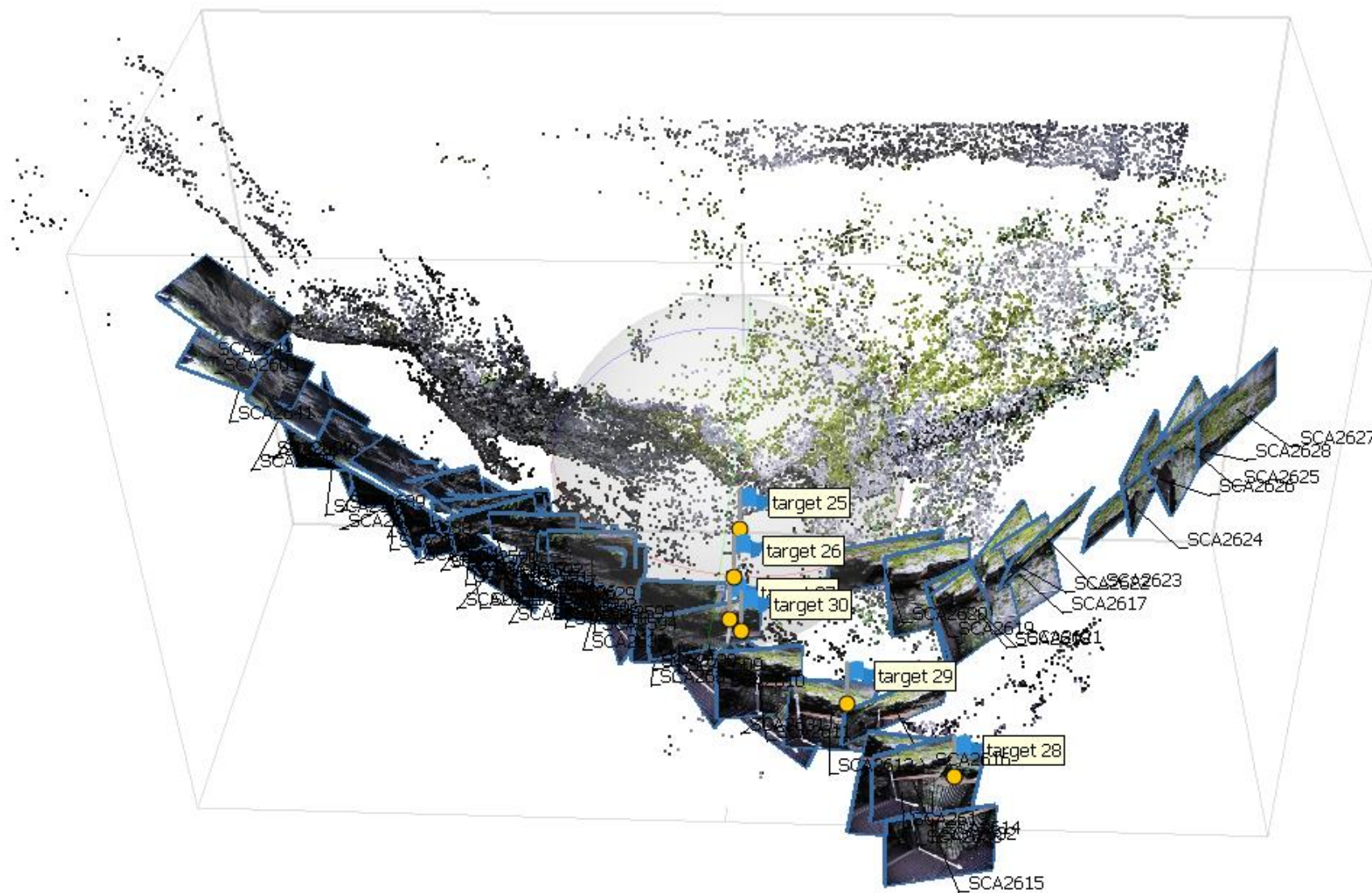
$$\text{Log}(x) = \text{Log}(H) - 0.664 + 0.1529 \text{Log}(V)$$

Vb(Size)	X(cm)	Ra(cm)
Average	59	88.9
Minimum	40	38.60
Maximum	76.84	153

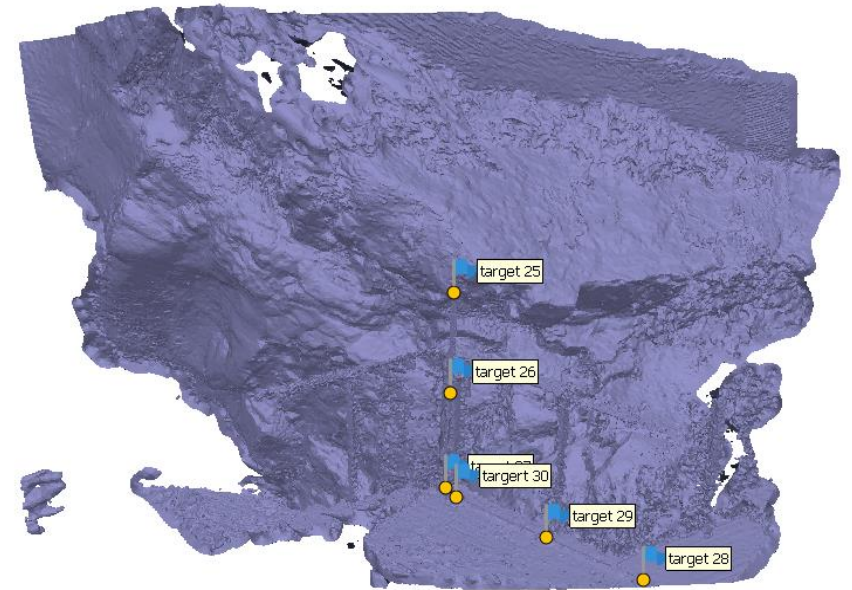
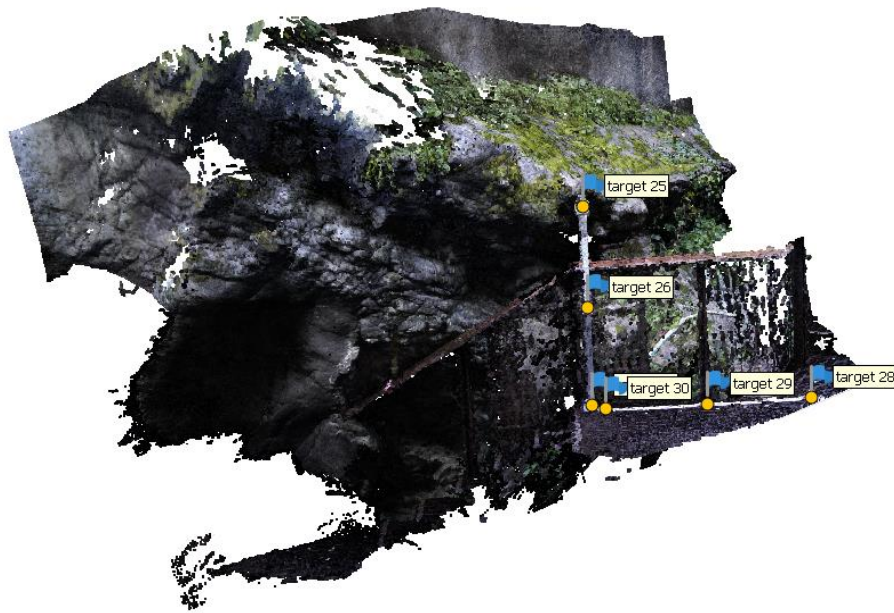
Table 24. Safety Distance of Blocks



Field Trip Photogrammetry



3D models of the Rock Surface





Families Location



Cell Fusion Parameters [Close]

Fusion algorithm
Kd-tree

Kd-tree cells fusion parameters

Max angle 10.00 deg.

Max relative distance 0.300

Facets

Max RMS 0.010

Min points per facet 500

Max edge length 0.50

OK Cancel

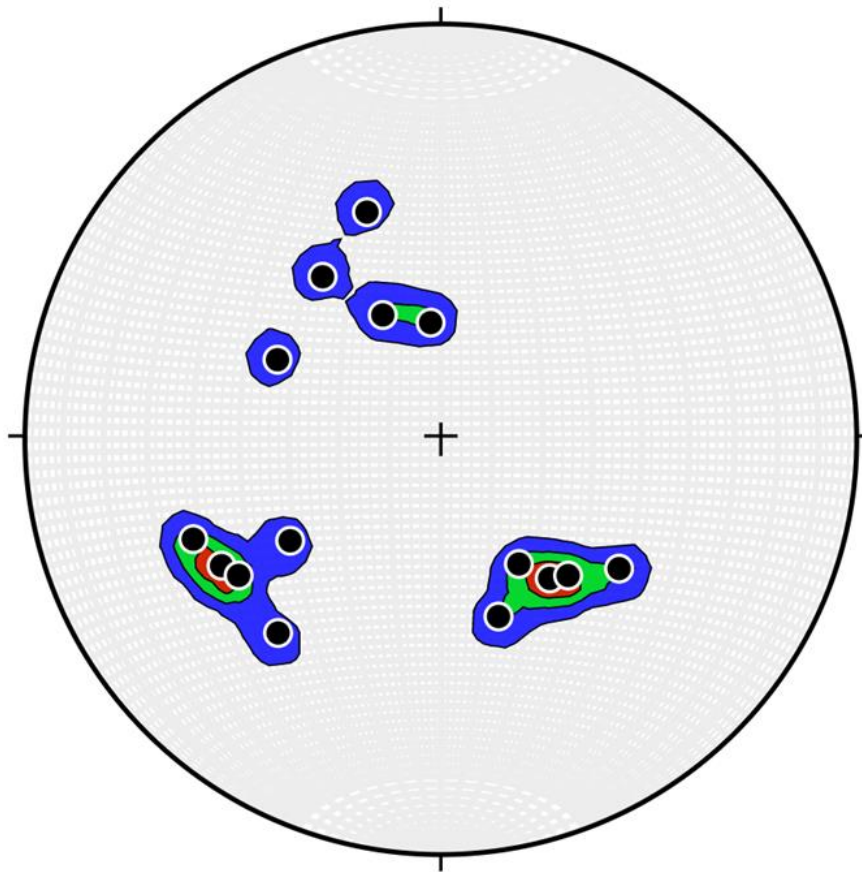


The dip and dip direction Software and In-situ

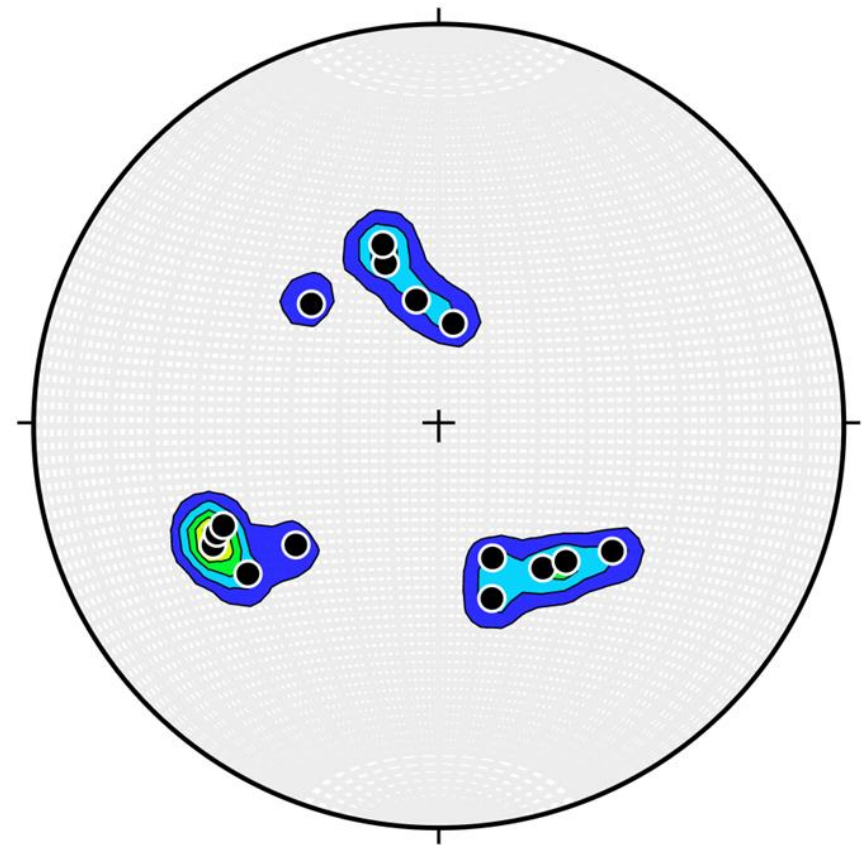
Family	In-Situ		Software		Difference		Average	
	dip	dip direction	dip	dip direction	dip	dip direction	dip	dip direction
Family 1	29	338	29	328	0	10	0,4	2,6
	37	343	37	342	0	1		
	36	324	35	322	1	2		
	38	317	37	317	1	0		
	44	306	44	306	0	0		
Family 2	53	62	54	68	1	6	0,6	7,4
	38	50	36	56	2	6		
	51	64	51	60	0	4		
	50	52	51	40	1	12		
	49	65	49	56	0	9		
Family 3	21	188	23	175	0	12	1,2	5,2
	26	170	27	155	1	10		
	35	162	40	144	3	4		
	39	163	48	162	1	0		
	36	134	36	116	1	0		



Stereograms of Software and In-situ Results



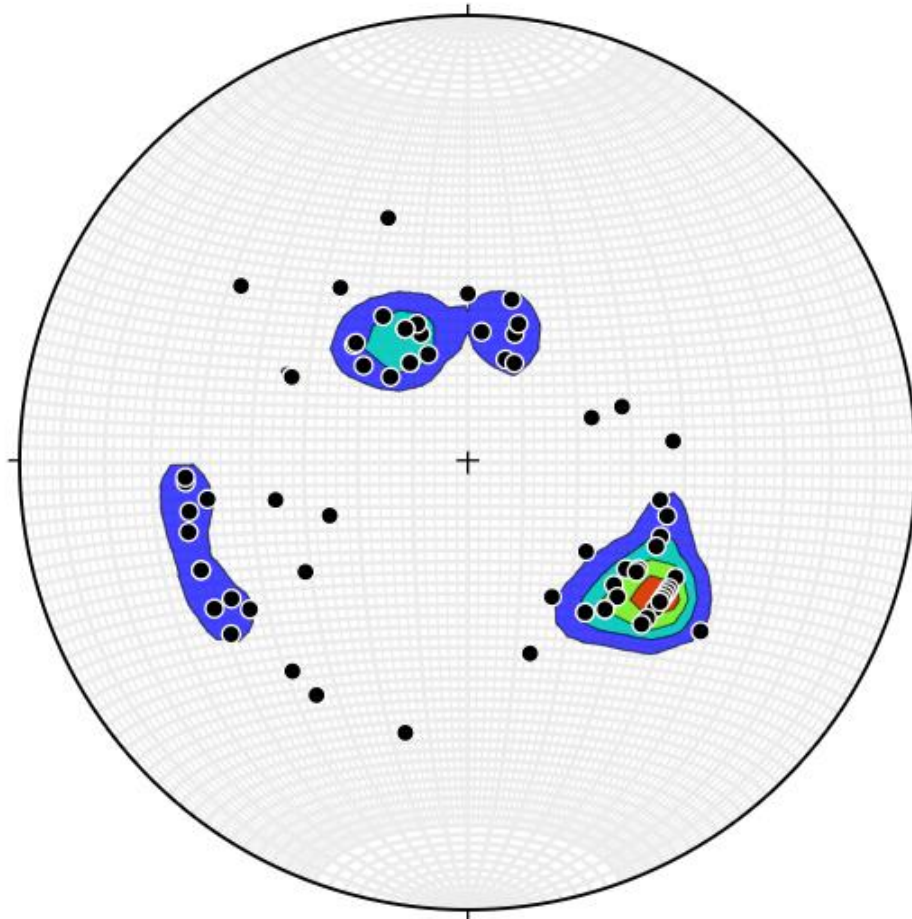
The Cloud Compare Results



The In-Situ Results



Stereograms of Global Results

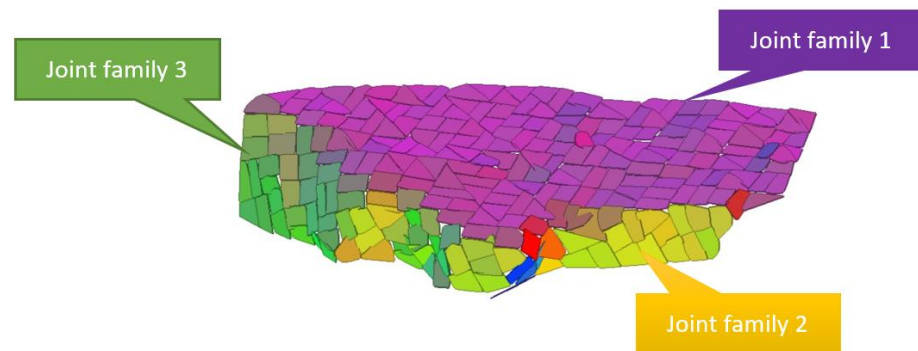
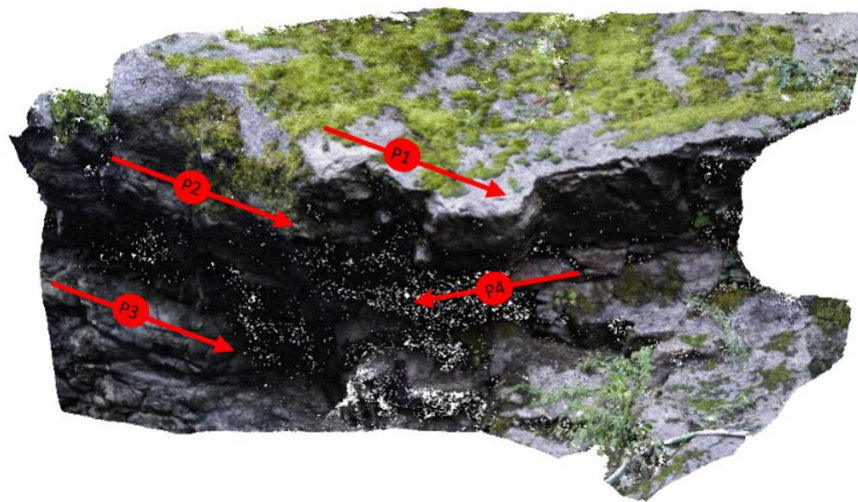


All Results from Software and In-Situ



Field Trip Geophysics

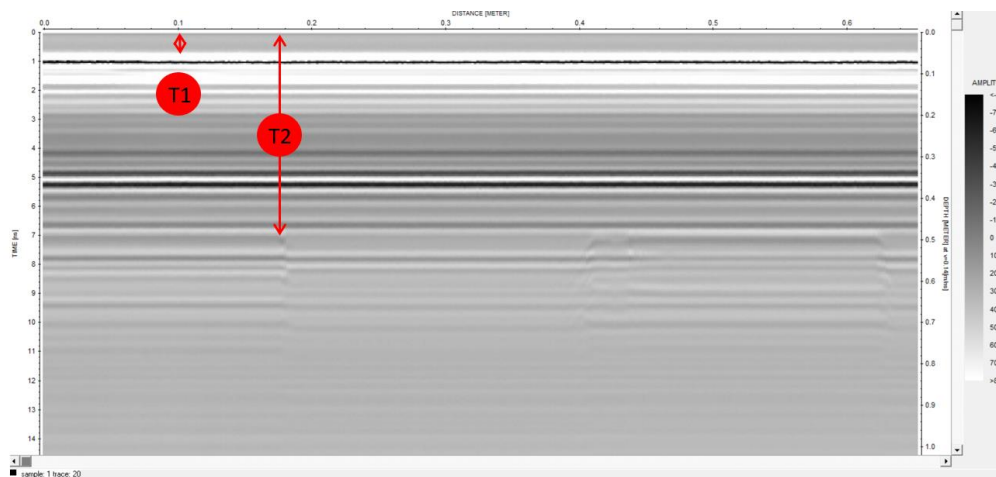
- Ground-penetrating radar(GPR)
- IDS (3GHz) GeoRadar





Velocity test

Velocity is estimate using distance over time, during this process, we main use the metal shield to create the reflection with opposite polarity





Time calibration

To calibrate the time scale so the time open the transmitting antennas is equal to 0

Bandpass filter

limit having sparse information ,select the range of frequencies

Gain

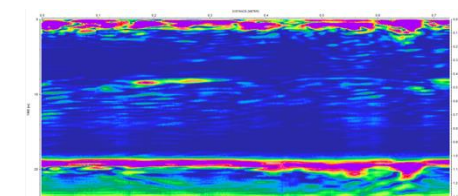
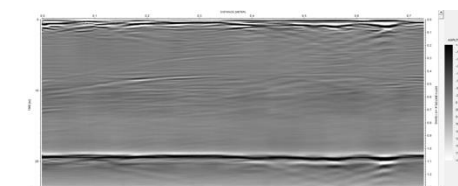
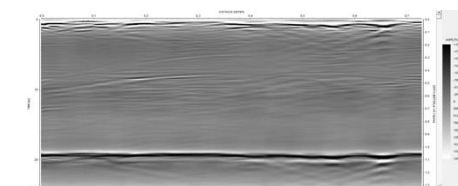
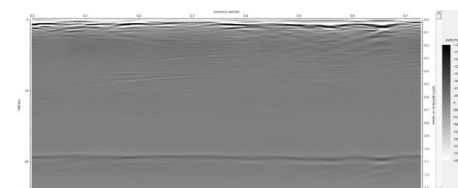
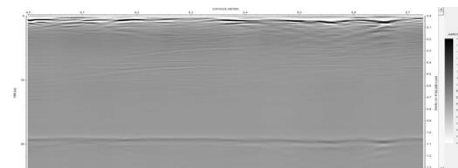
to amplify the defraction, to make the signal visible with the same amplitude with the background signal.

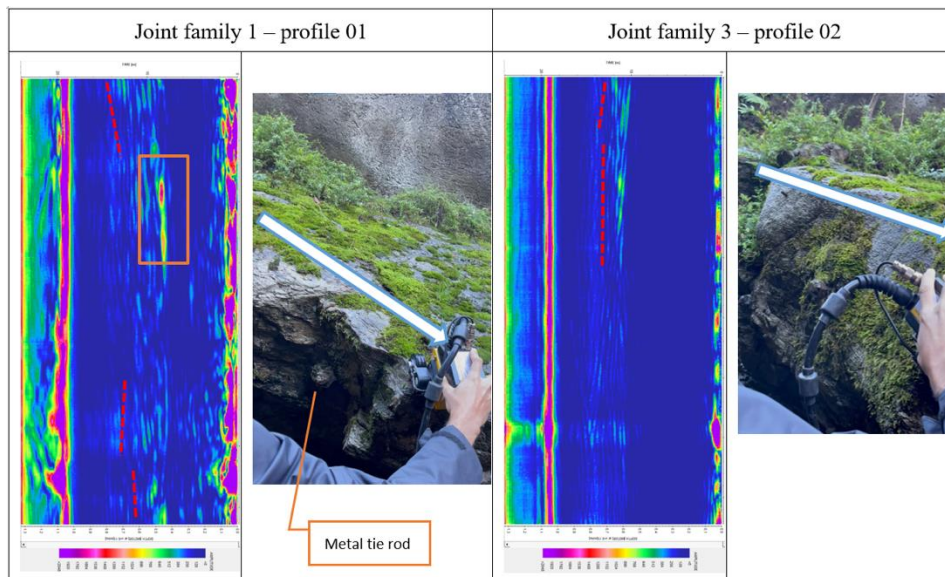
Background removal

Highlight the shallow signal which mask by background signal.

Envelope

remove the negative amplitudes from the profile by transforming waves with negative and positive oscillations into positive signals,



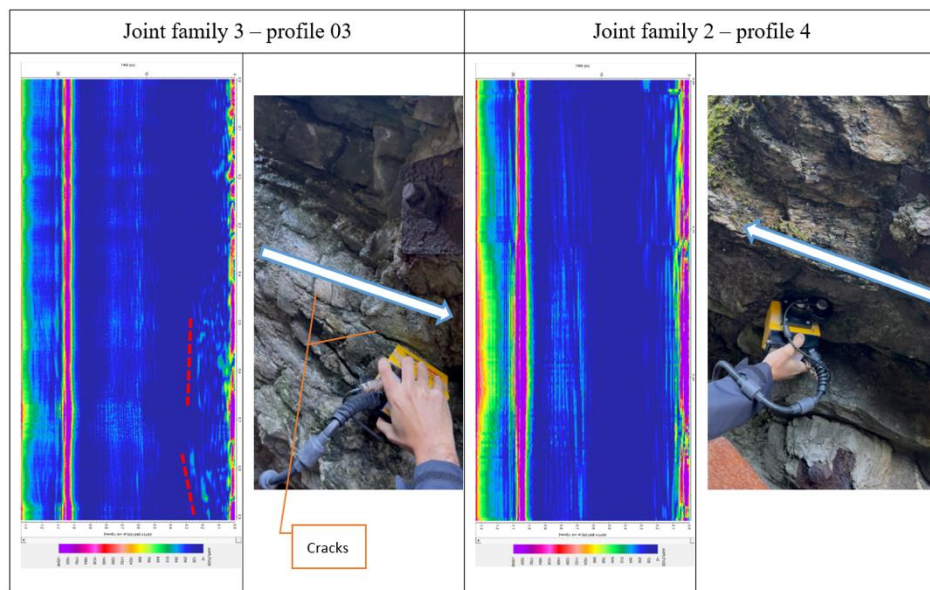


Profile 01 – family 1

- 1.Strong signal in depth 50cm shows a metal tie rod.
- 2.moderate joint-persistency, as discontinuities are quite far from each other.

Profile 02,03 – family 3

- 1.Not so strong signal in depth 15cm,maybe a metal or inner fracture.
- 2.a moderate joint-persistency, since discontinuities are short and far from each other



Profile 04 – family 2

- 1.signal is more horizontal and in low value because located in a less exposed place.
- 2.a low joint-persistency, since no evident discontinuity inside the rock.



A joint-persistency factor p is used for estimating our joint families:

$$p = \frac{L_j}{L_j + L_r}, L_j: \text{joint length}, L_r: \text{gap between joints}$$

Joint family	Estimated joint-Persistency (%)	Average joint-spacing (cm)
JF1	66%	5
JF3	68%	10

This parameter tells us the possibility of forming a line of crack inside the structure.

Just moderate joint-persistency.

In this project, a combination of various techniques and finally we finish our field trip.



Thank you!