

Design Theory for Secured Drapery

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Which is the right kind of mesh?
Which is the cooperation between nails and mesh?
Which is the right density of nails?



Top Wire Rope Cable



Cable Anchor



If the target is missed then the intervention is ...

NOT feasible and

NOT safe

for the contractor and the client



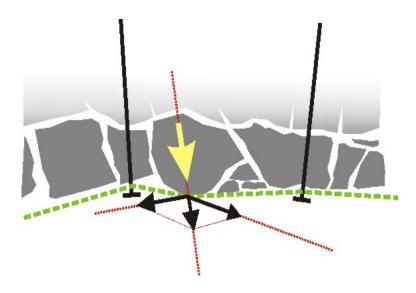


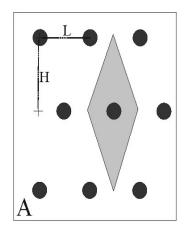
The superficial stability analysis is totally different for the rock than for the soil. This presentation is for the rock facing stability analysis.

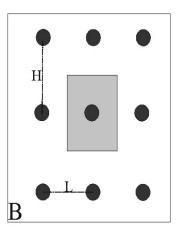












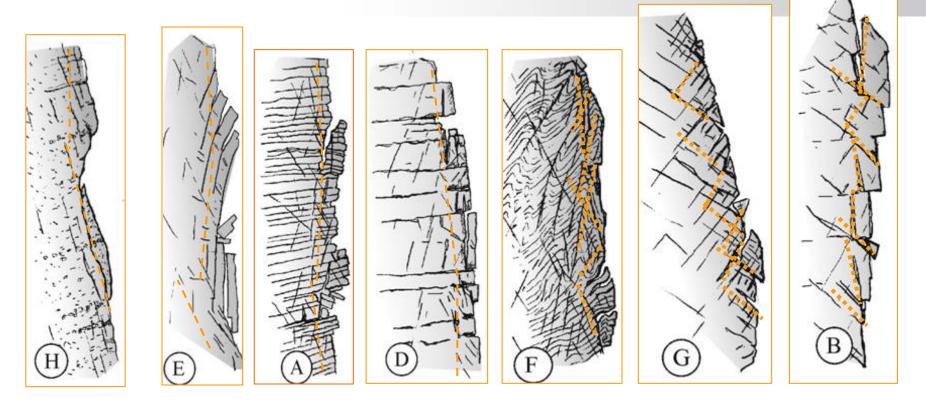
The stresses on the system must be controlled looking for the equilibrium between the stabilizing contribution offered from the various components of the system. This happens changing

- Spacing and resistance of the anchorings
 - 2) Tensile strength and stiffness of the wire/cable mesh

It is evident that the mesh can cooperate with the anchorings only if reacts to the pressure with the minimum deformation

(high stiffness).





The surface of the rock mass is a loose zone of a certain thickness.

On this zone there are sets of joints dipping towards the slope which create unstable conditions.

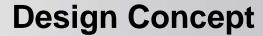




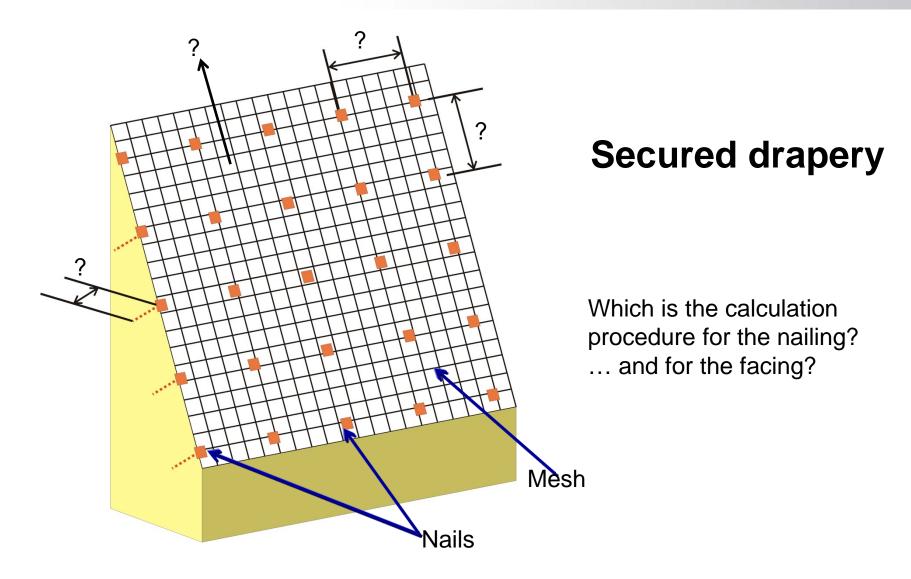
MACRO 1 software is a simple design approach for secured drapery system, which combines the field experience of geologists and engineers on one hand, and the results of full scale drapery field tests on the other.

The calculation procedure allows for determining both the ultimate limit state (verification of breaking loads of the system components), and serviceability limit state (maximum permissible deformation of the facing).





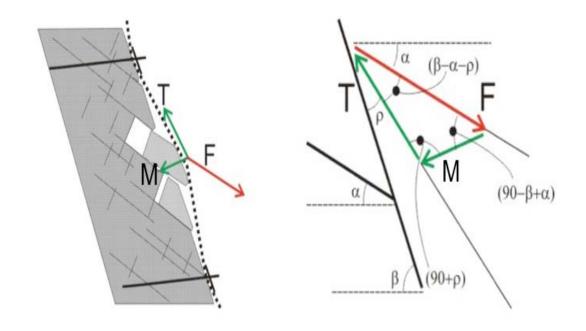












Equilibrium Design Theory

F= Forces developed by the block sliding

T= Mesh tensile resistance from secured by the upper anchors

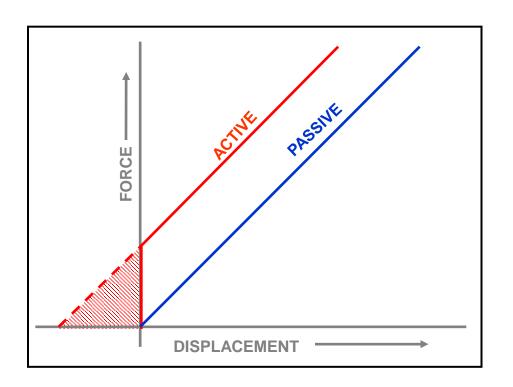
M= Punch resistance from the mesh under the block sliding





Design Concept

Secured drapery



Case (1) ACTIVE systems:

The force Rb acts before the driving forces overcome the resisting ones. *Rb acts against the driving forces:*

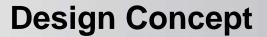
$$Fs = \frac{Re \, sisting \quad forces}{Driving \quad Forces \quad -Rb}$$

Case (2) PASSIVE systems:

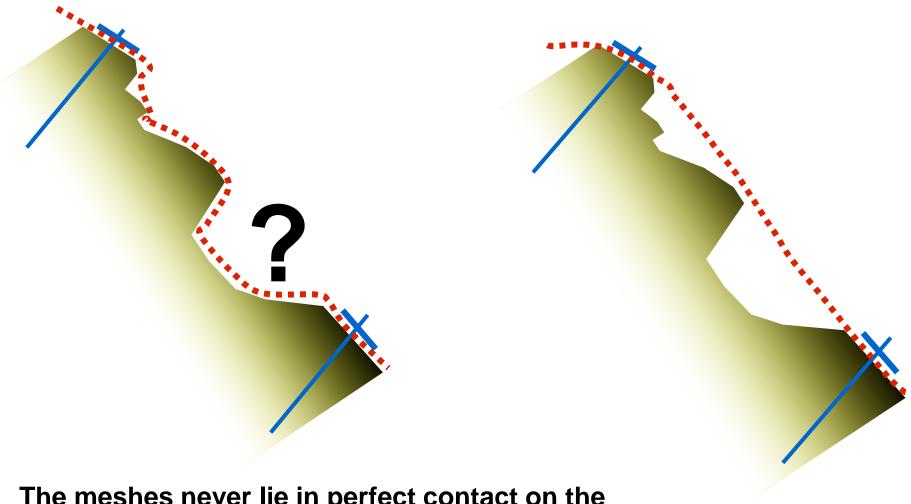
The force Rb acts after the rock mass becomes unstable. Rb acts with the resisting forces and not against the driving forces:

$$Fs = \frac{\text{Re sisting Forces} + (Rb)}{Driving forces}$$



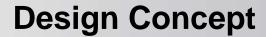






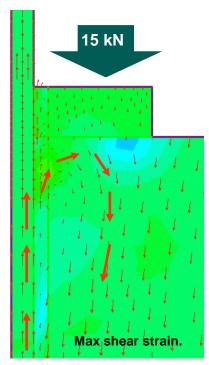
The meshes never lie in perfect contact on the ground surface





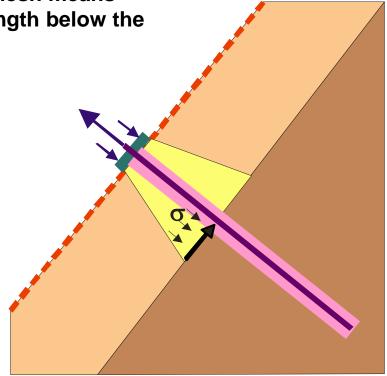






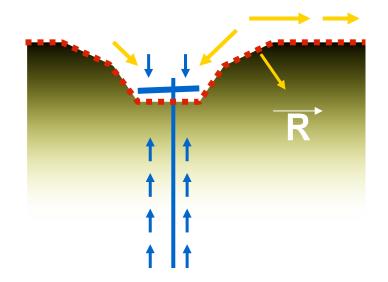
When using nails, pre-stressing down plate and mesh means short circuit strength below the plate.

No pressure is developed on sliding plane







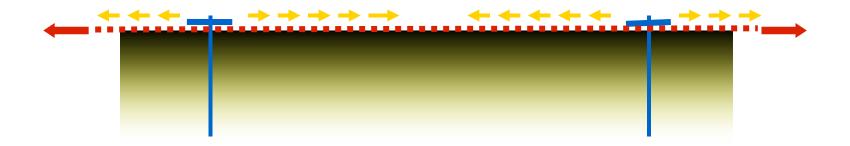


The meshes pushed down in the hollows can develop localized forces only (R).

The modulus of R is absolutely negligible





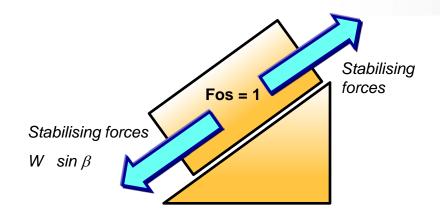


On planar surfaces the meshes can develop tangential forces only

That is why, even if the mesh is pre tensioned, no one pressure acts on the ground.



Design Concept

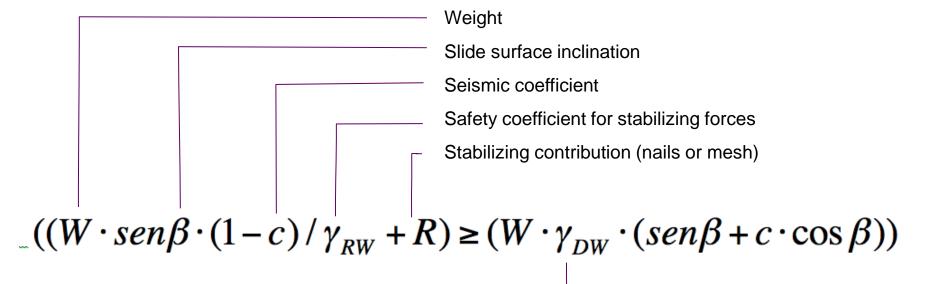


Passive Design Concept

Conceptual solution:

Stabilizing forces + R > Driving forces
$$\gamma_d$$

Safety coefficient for driving forces



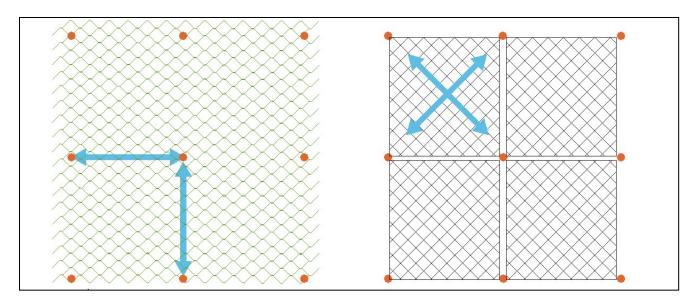




Load transfer

WIRE MESH PANELS

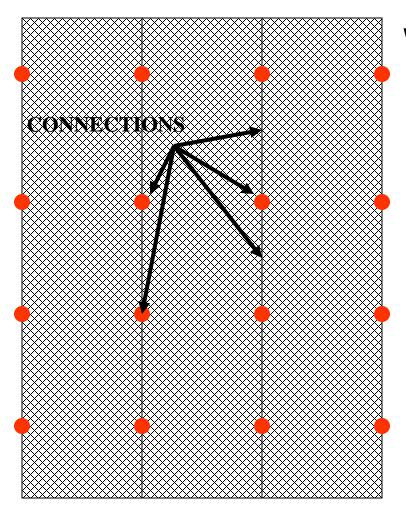
CABLE MESH PANELS



The load is transferred as per the mesh fabrication patern to the closest anchorings. The load is transferred to the anchorings along diagonal directions



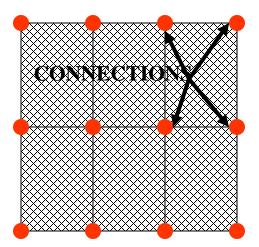
Border conditions



Wire mesh rolls: important the connection along the longitudinal borders

1

Importance of the benefit offered from RockMesh HR provided with longitudinal cables Cable mesh panels: the load is transferred along the "diagonal directions" and the border cable doesn't improve the system behaviors





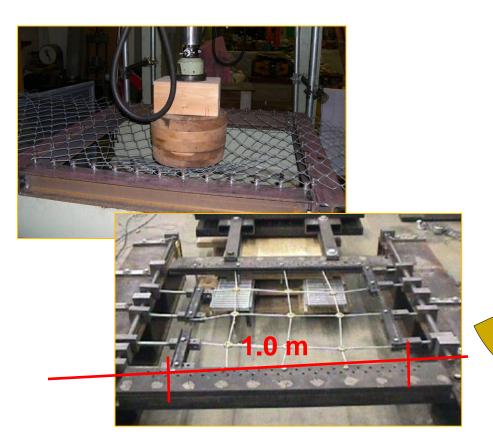


Testing is required to determine the behaviour of the mesh/panel with the action of the rocks.

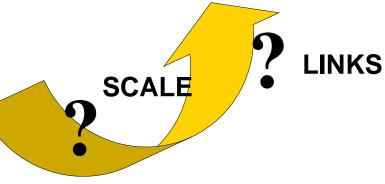




Scale Tests







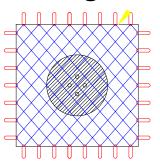




Large Scale Tests

Assessing the rigidity of deformity products

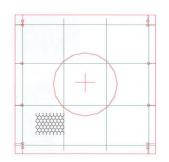
HEA Panel
wire Ø10, Mesh 400x400







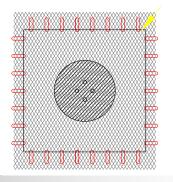
RockMesh
Mesh 8x10/Ø 3.00 Galfan







Chain-Link wire A.R., Ø 3.00



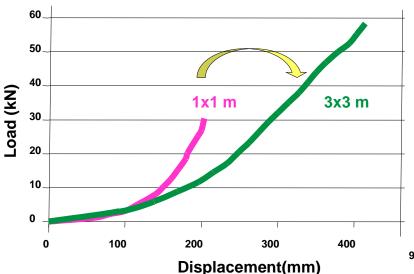






Large Scale Tests

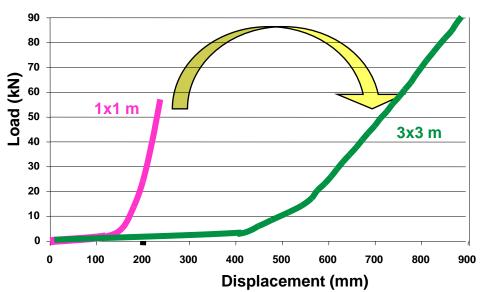
Double Twisted Mesh - 3.00 mm



Importance of full scale tests.

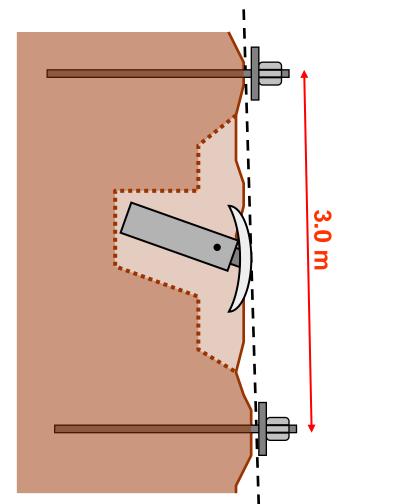
Chain-Link Mesh - 3.00 mm

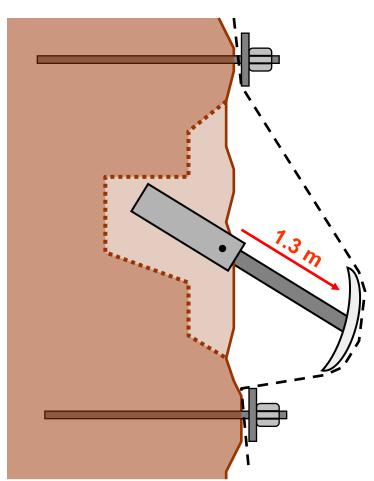
Comparison between
Scale tests (1x1 m)
Real Scale tests (3x3 m).





Pont Bozet (AO – Italy) - mesh field test





Large scale tests 3 x 3 m - Sample restrained on 4 points



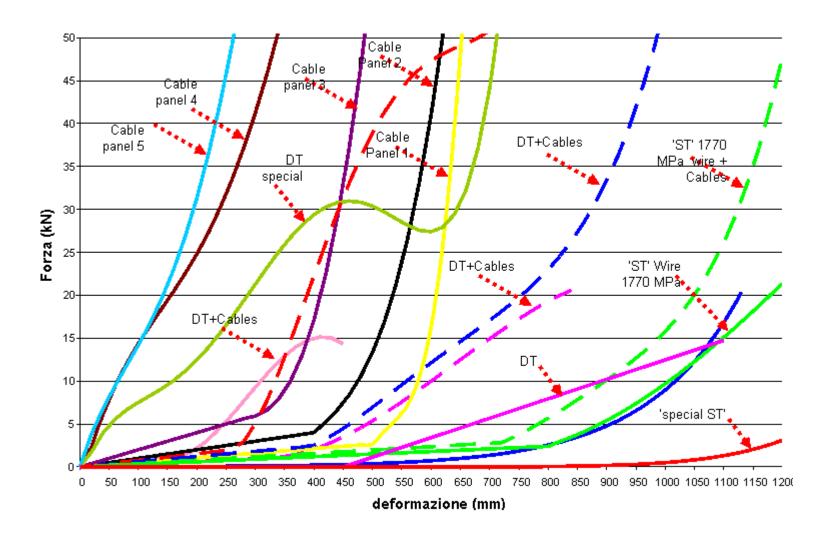


HEA Panel & DT mesh



Panel HEA with Knots

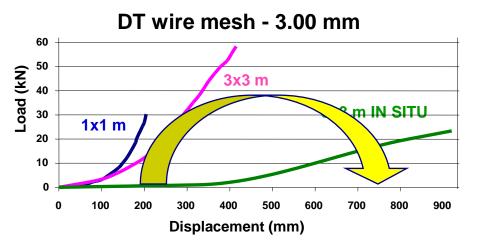
Pont Bozet (AO)
Test 21/02/2007
HEA Panel 300/10







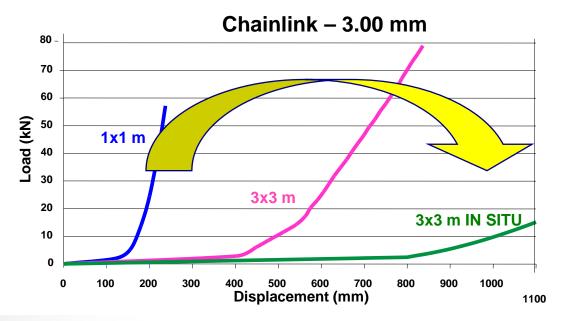




The importance of the test in the real size with real connections.

Comparation:

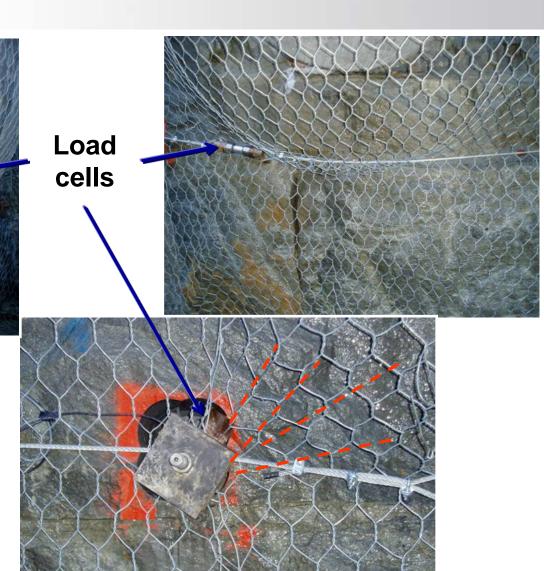
- Small size (1x1 m)
- Large Size (3x3 m) in Lab,
- Real Size (3x3 m) in situ

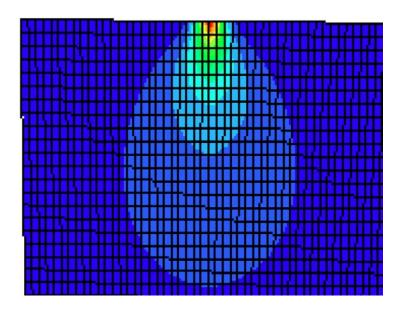






Tests have
demonstrated that high
stress is generated at
anchor point and steel
wire rope cable was the
best solution for a
strong connection.





Modeling of the forces at the mesh anchor point.

The highest stress is below the anchors which are stiff restrain. Between the anchors the stress is low even if the mesh has a high tensile resistance (i.e. 170 kN/m). Actually the tensile strength has no importance if the mesh is not stiff.

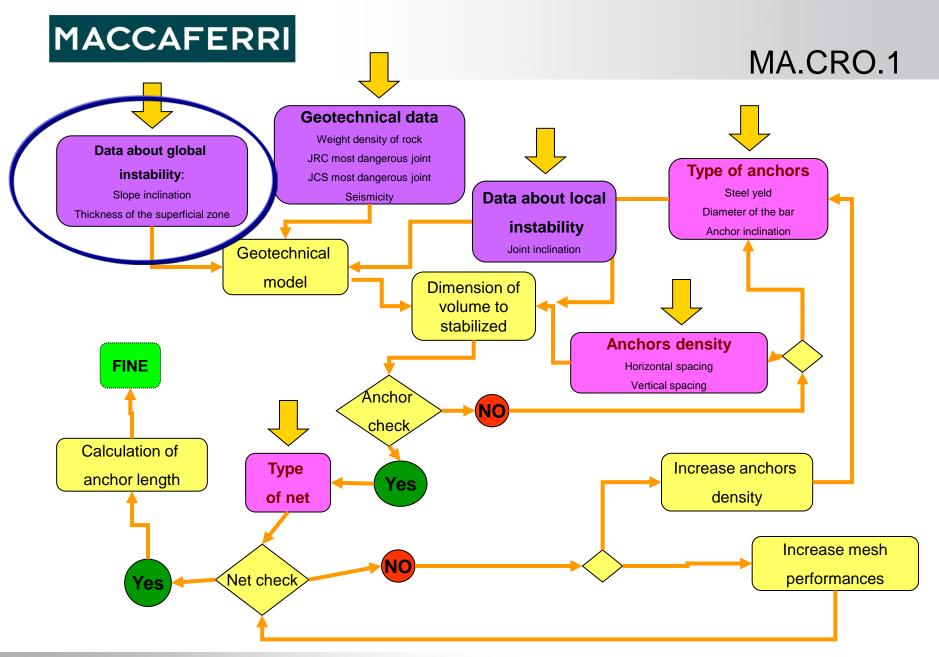




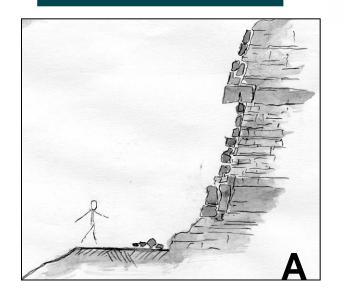
MACRO 1

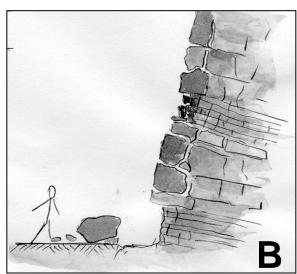
a simple design approach for secured drapery







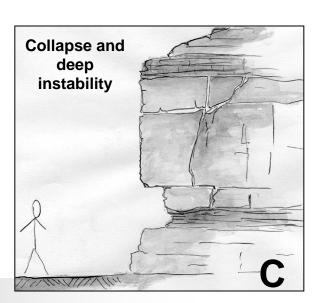


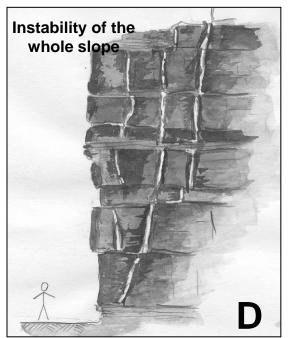


Global Instability. It needs a geomechanical

survey

The surface instability is relates to external and weathered rock surfaces. The superficial instability doesn't affect the overall stability of the slope.





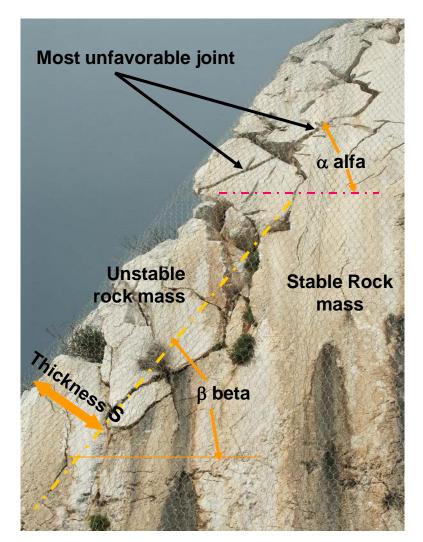




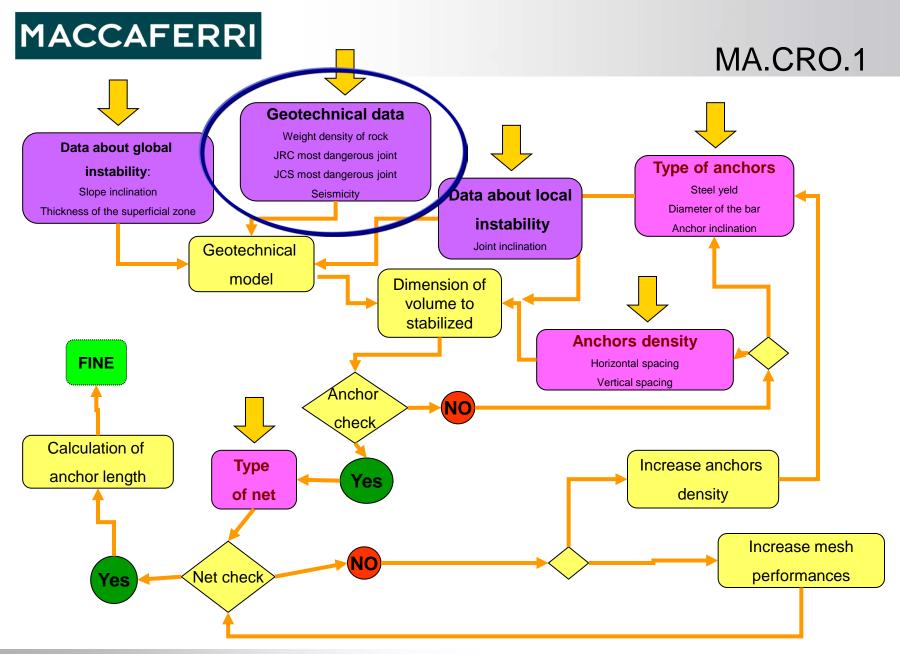
Software concepts

S = The average thickness of the loose rock mass

 β = average inclination of slope surface



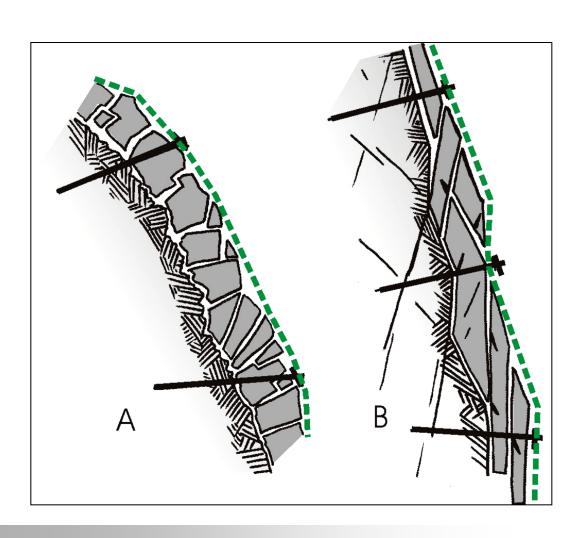








Modelling of the surficial portion of the rock mass



Based on practical experiences, the surficial portion of the rock mass can be traced back.

- A) Pseudo-continous behavior: the surficial portion of the rock mass is highly divided. The wire mesh develops a relevant confinement action. The anchorings are only occasionally affected by shear stress.
 - B) Discontinous behavior: the surficial portion of the soil mass is lightly divided. The wire mesh develops a local confinement action. Anchorings are frequently subjected to shear stress.

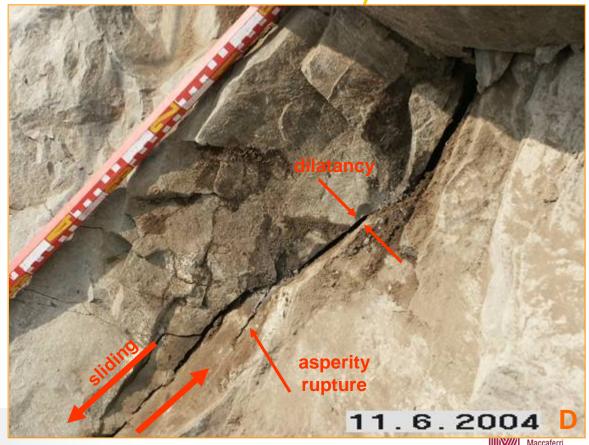


The importance of natural joints in the rock mass



JRC most dangerous joint JCS most dangerous joint

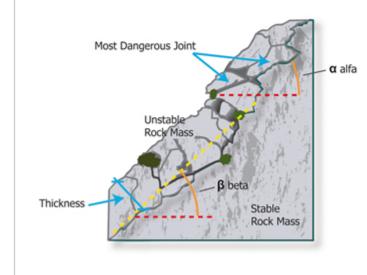




Compressive Strength (JCS)

JCS is the Unconfined Compressive Strength of the most dangerous joint set. JCS is measured by Schmidt hammer. In a rough way we can take that: JCS = 1/3 UCS (Unconfined Compressive Strength).

JCS increases the resistance of the anchorage under shear stress. The software corrects the JCS in order to take into account the scale factor.



Rock	Compressive Strength (MPa
Cementing conglomerate	70 - 100
Cementing sandstone	75 - 160
Siltiti	50 - 180
Mudstone	50 - 180
Sandstone	2 - 150
Marl – marlstone	25 - 90
Dolomite	60 - 300
Chalk	15 - 30
Granite	95 - 230
Porphyry	100 - 230
Dolerite	220 - 320
Tuff	3 - 100
Andesite	75 - 300
Basalt	100 - 350
Gneiss	80 - 160
Marble	60 - 230
Serpentine	20 - 130
Phyllite	20 - 80

Compressive Strength (Minimum value) = 10

Ok



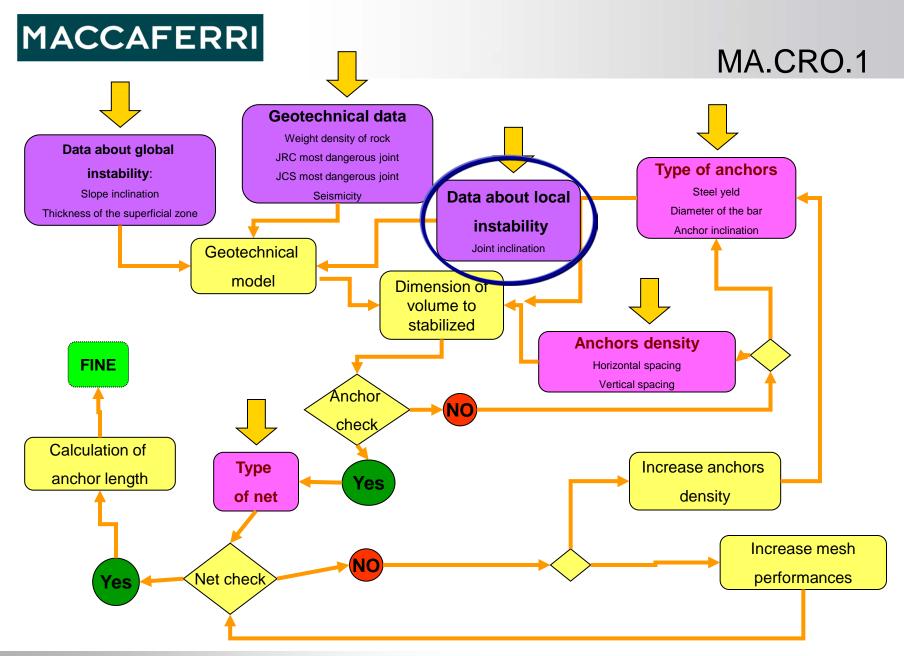
Roughness

Roughness of the Most Dangeorous Joint set is measured by Barton "comb". Its value range is between 0 t (smooth) to 20 (very rough joint). If the joint is weathered or day covered, it s better to take low values of Roughness.

Roughness increases the resistance of the anchorage under shear stress.

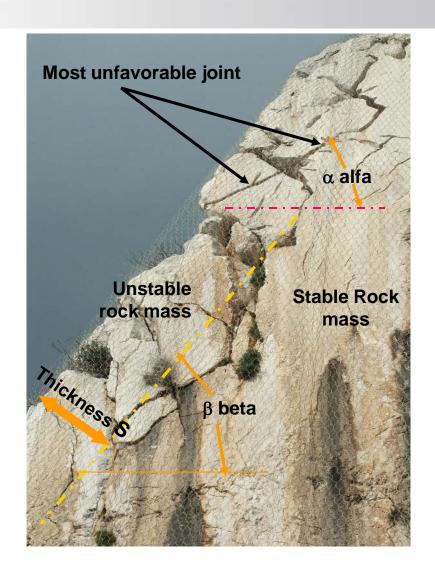
Profile	Roughness
	0 - 2
	2 - 4
	4 - 6
	6 - 8
	8 - 10
~~~~	10 - 12
	12 - 14
	14 - 16
	16 - 18
	18 - 20
0 5 10 cm	
Roughness = 0.1	.]
Ok	



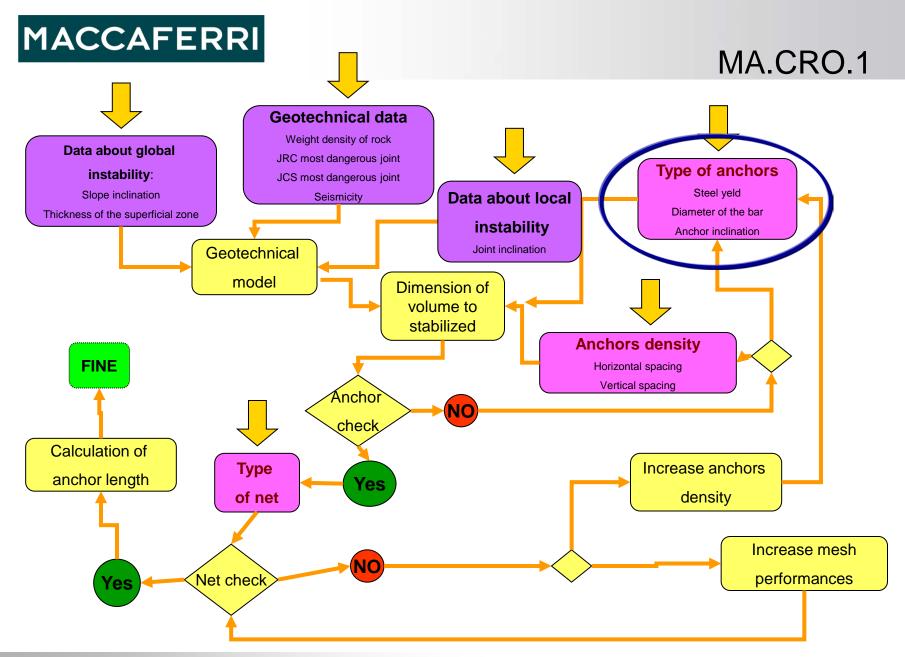




 $\alpha$  = Angle of the most unfavorable joint







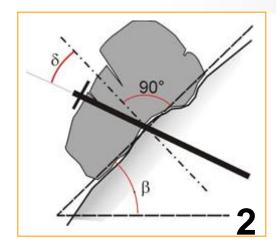


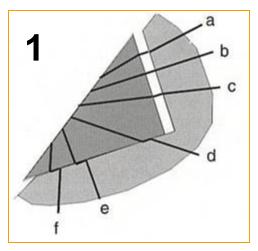
Nominal diameter of the anchorage bar.

Type of the bar with yield strength



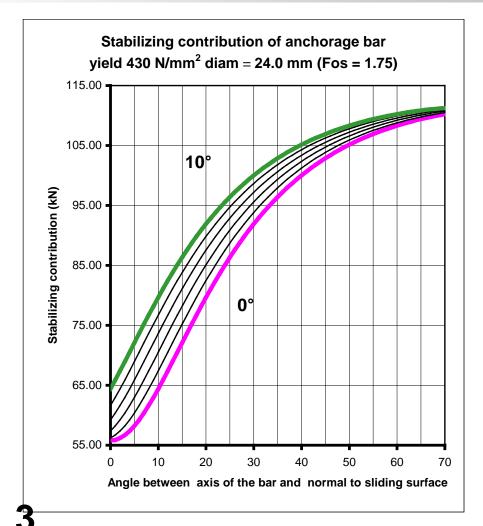




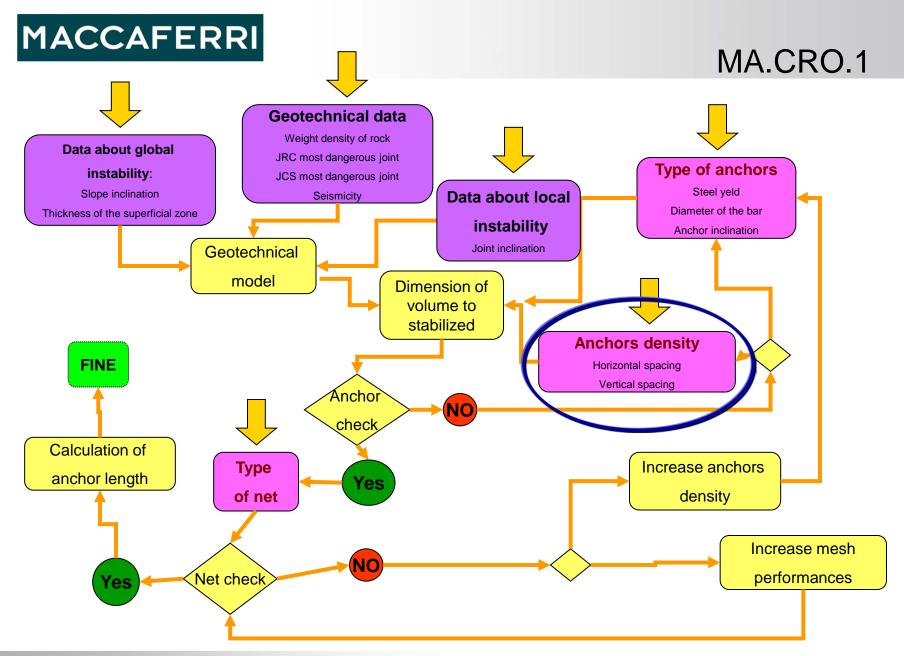


Bar subjected to pure traction (case "b"): the joint dilatency does not affect the resistance contribution due to the bar.

Bar subjected to pure shear ( case "e" ): the greater the joint dilatancy, the higher the resistance contribution of the bar.







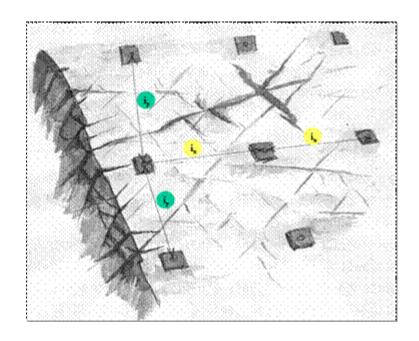


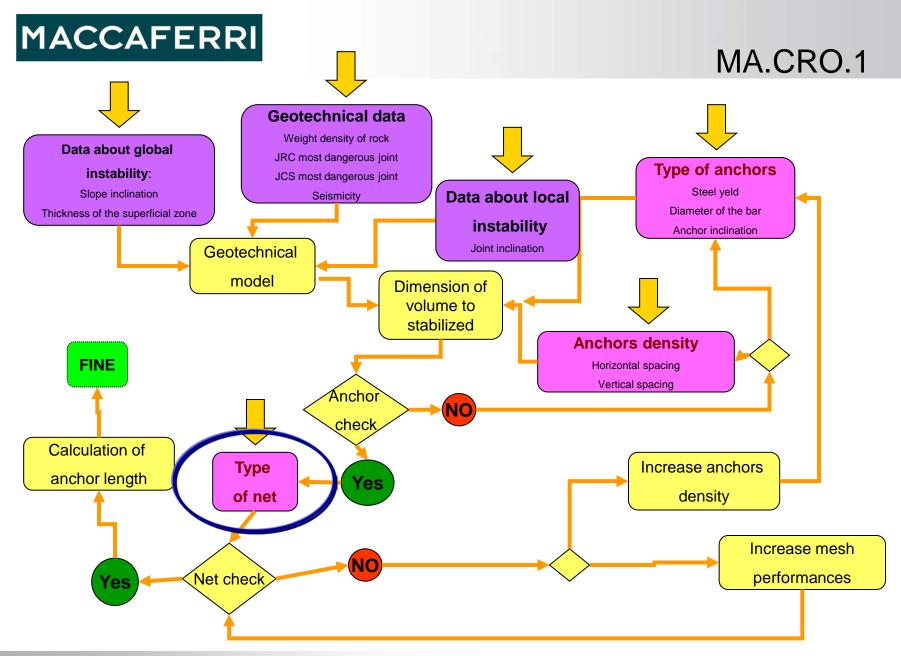
Spacing horizontal ix between the anchorages.

It is better than ix doesn't differ too much from iy.

Vertical spacing iy between the anchorages.

It is better than iy doesn't differ too much from ix.







Mesh DT PVC 8x10 / 2.7 mm

Mesh DT 8x10 / 3.00 mm

RockMesh HR 30

RockMesh B600

RockMesh B900

HEA panel 300 mm / 10 mm

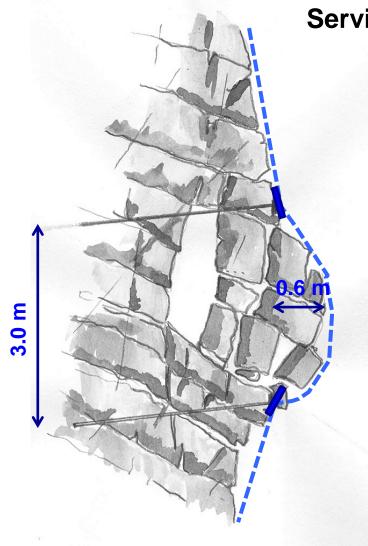
HEA panel 400 mm / 10 mm

HEA panel 300 mm / 8 mm







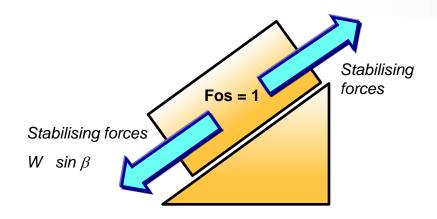




The acceptable maximum bulging must be assumed taking into account of maintenance problems & geotechnical conditions.

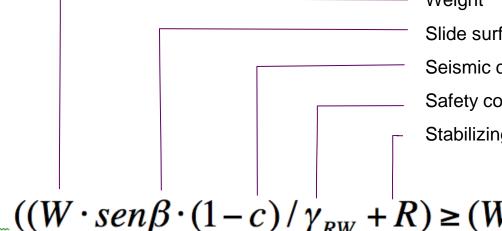


## **Design Calculation**



Conceptual solution:

Stabilizing forces + R > Driving forces 
$$\gamma_d$$



Weight

Slide surface inclination

Seismic coefficient

Safety coefficient for stabilizing forces

Stabilizing contribution (nails or mesh)

$$((W \cdot sen\beta \cdot (1-c)/\gamma_{RW} + R) \ge (W \cdot \gamma_{DW} \cdot (sen\beta + c \cdot \cos\beta))$$

Safety coefficient for driving forces





 $\gamma_{\rm DW} = \gamma_{\rm MO} \ \gamma_{\rm OL}$ 

## **Design Calculation**

		uncertainties while determining the surficial instal	oility thickness s ;	1.20 - 1.30
		uncertainties in the unitary weight;		1.00 - 1.05
		uncertainties related to the rock mass weathering	and erodibility;	1.00 - 1.05
$\gamma_{\rm RW} = \gamma_{\rm THI}$	γwG	γвн	(for stabilizin	ig forces)

(for driving forces)







### MACRO 1 Reinforced System

Rock and Soil Slope Protection Design Software



Client: / pag. 1 of 2

### Project Information

Title: Description: Number:

#### Input

Client: Designer:

#### Rock Slope

Slope inclination [°]	В	65
Slope total height [m]		
Thickness of the surficial instability [m]		0.5
Density of the rock mass [kN/m3]		27
Assumed length os plasticization n the unstable rock mass [m]		0.3

#### Most Dangerous Joint

Inclination [º]	25
Compressive strength [MPa]	10
Roughness	0

#### Seismic Acceleration

Horizontal seismic coefficient	0.14

#### Mesh

Mesh Type	DT 8 x 10 Ø 2.7	
Mesh ultimate tens	sile resistance [kN/m]	60
Maximum displacer	ment acceptable [m]	0.6

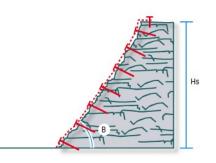
#### Anchor Bars

#### Geometry

Horizontal anchor spacing [m]	3
Vertical anchor spacing [m]	3
Inclination of bar to the horiozntal [º]	10

#### Anchor Type

Bar type	_bar	
Bar internal dia	meter [mm]	0
Bar external diameter [mm]		16
Thickness of corrosion crown [mm]		0
Bar yield stress (of steel) [MPa]		500
Rock-grout adh	nesion (bond stress) [MPa]	2.25



#### Safety Factor

#### Factors affecting the stabilizing forces

Uncertainty of the thickness of surficial instability	1.2
Uncertainty of the rock mass unit weight	1.01
Uncertainty of rock behavior and weathering	1.02
Control: Safety factor do reduce stabilizing forces	1.24

#### Factors affecting the driving forces

1.1
1.02
1.12
1.39

#### Mesh

Safety reduction for mesh resistance	2.5
Safety reduction for maximum displacement	2

#### Anchor type

Safety reduction for steel resistance	1.16
Safety reduction for rock-grout adhesion	2





### MACRO 1 Reinforced System

lock and Soil Slope Protection Design Software



Client: / pag. 2 of 2

#### * Results

Bar design check

Mesh design check

1.01 Satisfied

1.61 Satisfied

#### Serviciability design check

1.20 Satisfied

#### Bar design

Stabilizing forces [kN]	132.36
Slidind plane driving force [kN]	131.38
Ratio strength stress	1.01
Bar inclination from horizontal [0]	15.00
Minimum acceptable steel yield stress [N/mm2]	431.03
Control: Effective cross section of bar [mm2]	201.06
Sliding plane stabilizing forces per acnhorage [kN]	55.99
Minimum driling diameter (Nominal) [mm]	38.00
Anchor pull-out force from load on the mesh [kN]	28.89
Anchor pull-out force due to global instability [kN]	14.24
Maximum pull-out force (total) [kN]	28.89
Minimum bar lenght in stable rock mass [m]	0.30
Minimum lenght (bar) in the unstable rock mass [m]	0.60
Minimum total bar lenght (Nominal) [m]	1.20

### Mesh design Maximum tensile fi

Maximum tensile force in the cable [kN]	24.00
Maximum tensile stress within the mesh [kN]	14.88
Force-strength ratio	1.61
Potential unstable volume on joint - case A [m3/m]	0.00
Potential unstable volume on joint - case B [m3/m]	1.20
Potential unstable volume on joint - case C [m3/m]	0.15
Maximum rock vol that can slide between anchors [m3/m]	1.35
Maximum rock weight that can slide betweem anch [kN/m]	36.48
Sum of the driving forces acting on the sliding plane [kN/m]	22.45
Sum of stabilizing forces acting on the sliding plane [kN/m]	10.69
Punching forces acting on the mesh [kN/m]	22.67
Mesh deformation angle from horizontal [o]	9.46

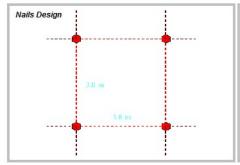
#### Serviciability

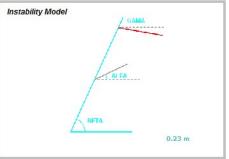
Serviciability	
Stabilizing forces [kN]	0.30
Slidind plane driving force [kN]	0.25
Ratio strength stress	1.20

#### Features of the instability

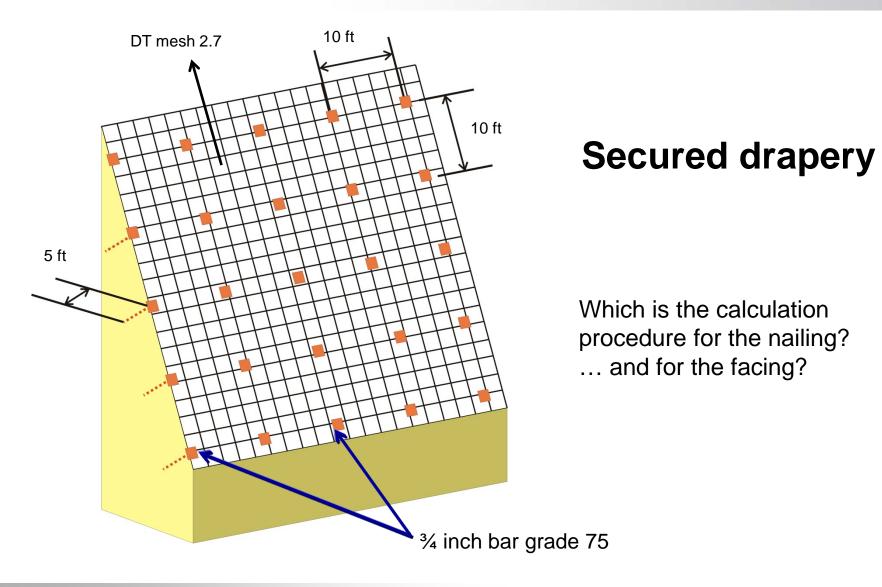
Tension on the averege slip surface [MPa]	0.01
Initial dilatance of the most dangerous joint [o]	0.00
Total unstable vol controlled by each anchorage [m3]	4.50
Total unstable weight controlled by each anchorage [kN]	121.50

#### Geometry













Even if the software allows a quick and simple calculation approach, onsite observations are always recommended to achieve a good design, with the ultimate goal of protecting property and the public.

