

Drilling

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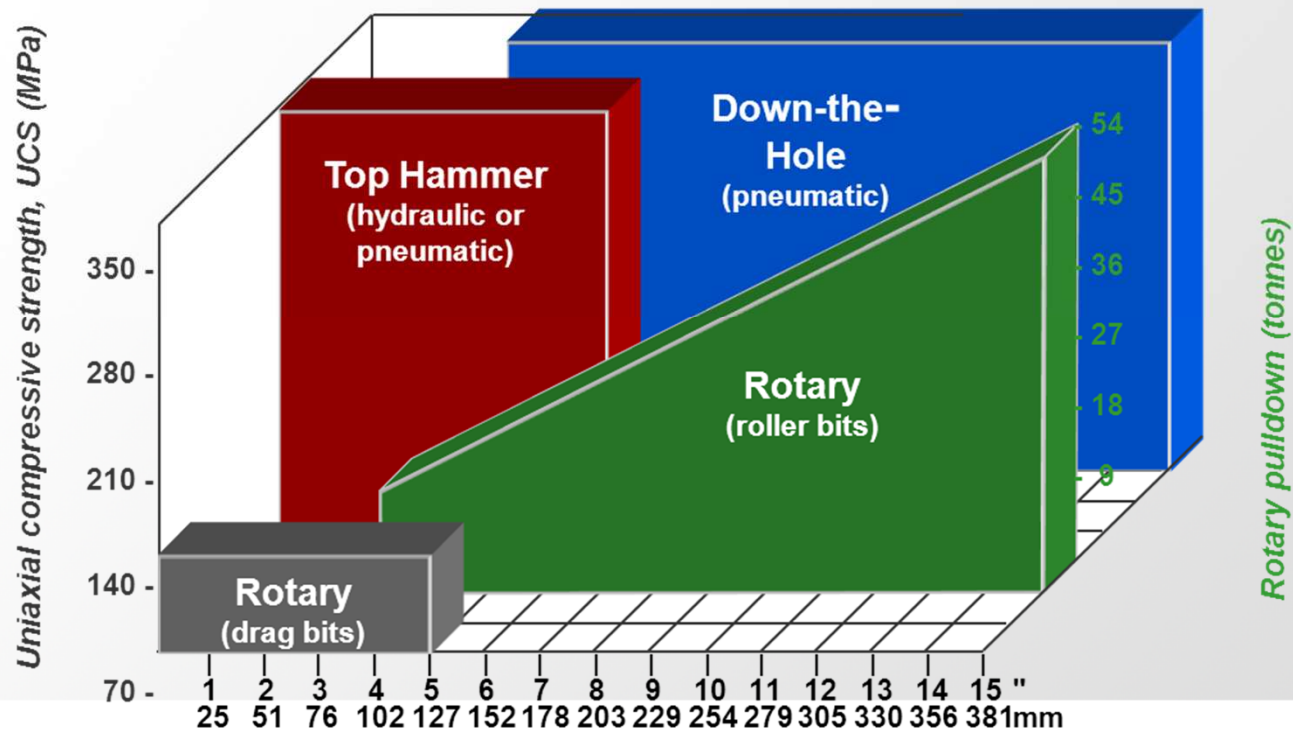


Agenda for drilling operations

- well planned operations and correctly selected rigs yield low cost drilling
- technically good drilling (good drill settings) and correctly selected rigs yields low cost drilling
- straight hole drilling yields safe and low cost D&B operations



The most common drilling methods in use



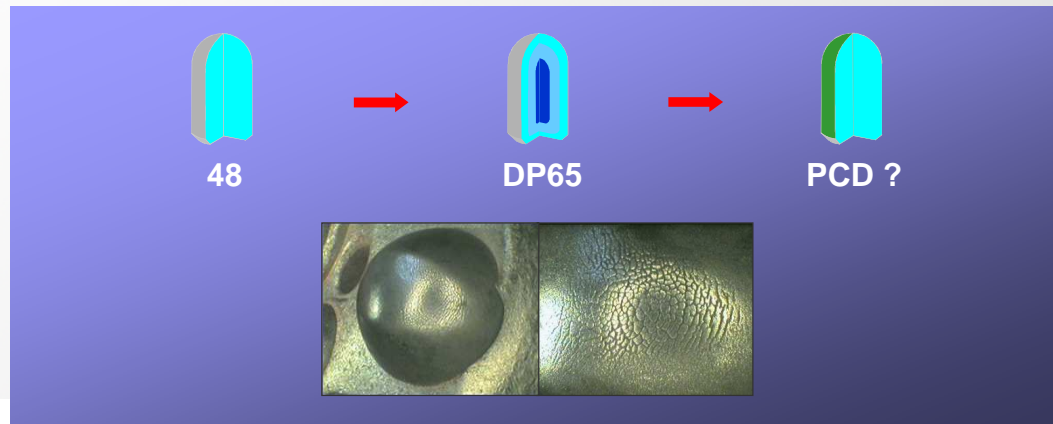
Drilling consists of a working system of:

- bits
- drill string
- boom or mast mounted feed
- TH or DTH hammer / Rotary - thrust
- drill string rotation and stabilising systems
- powerpack
- automation package
- drilling control system(s)
- collaring position and feed alignment systems
- flushing (air, water or foam)
- dedusting equipment
- sampling device(s)



Guidelines for selecting cemented carbide grades

- avoid excessive button wear (rapid wearflat development)
 - => select a more wear resistant carbide grade or drop bit RPM
- avoid button failures (due to snakeskin development or too aggressive button shapes)
 - => select a less wear resistant or tougher carbide grade or spherical buttons
 - => use shorter regrind intervals



Selecting button shapes and cemented carbide grades

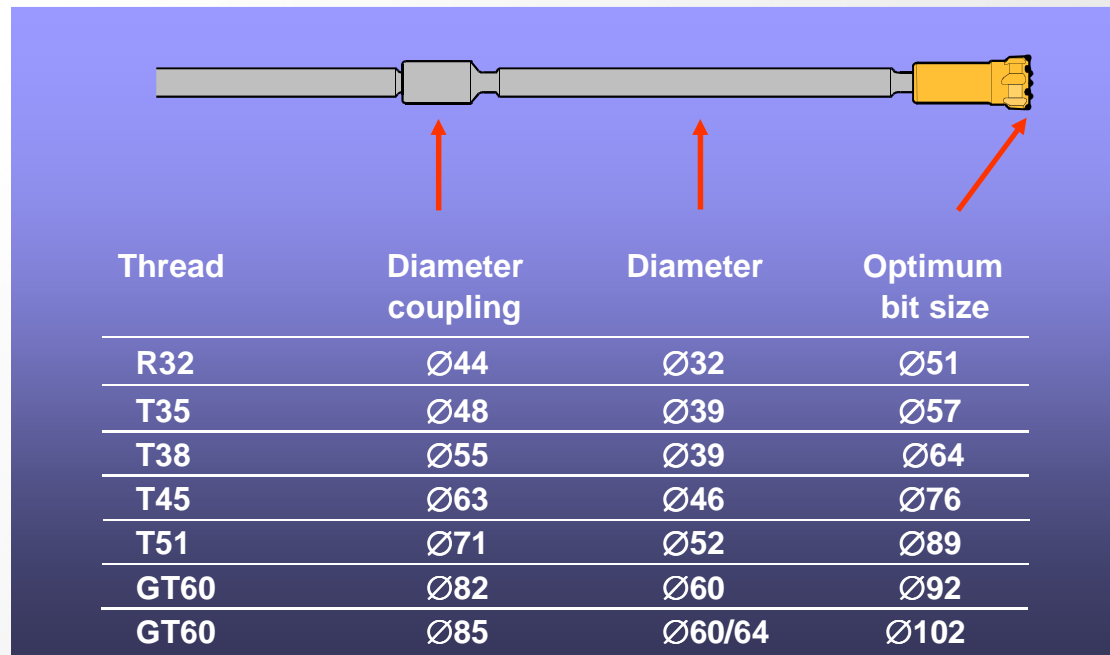


Spherical buttons
DP65
S65

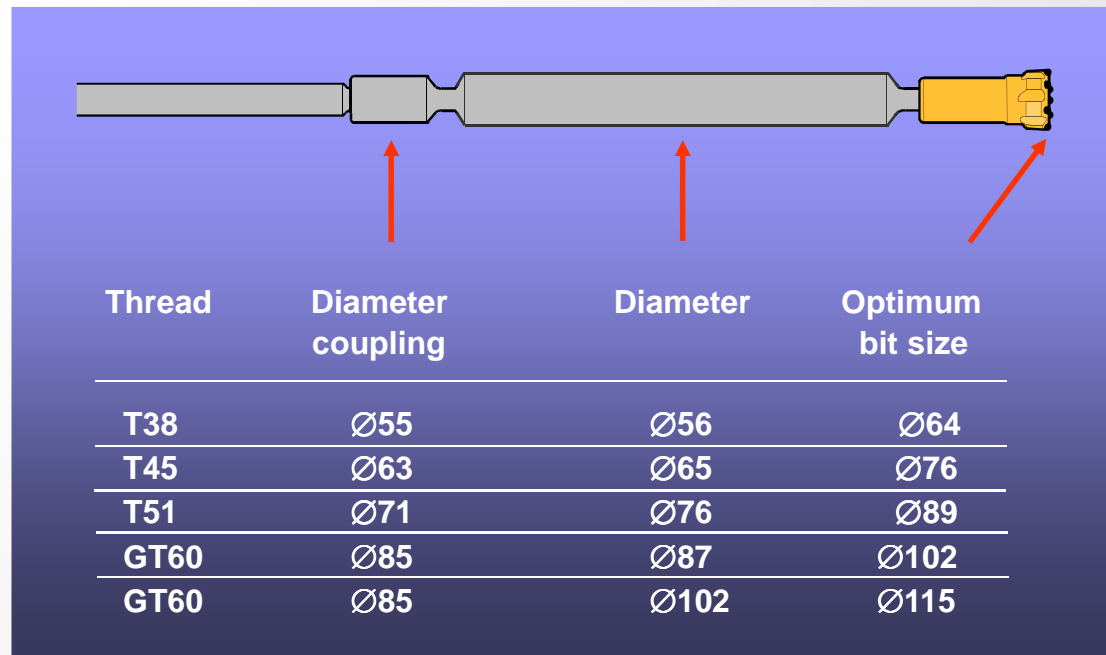


Robust ballistic buttons
48
R48

Optimum bit/rod diameter relationship for TH



Optimum bit/guide or pilot (lead) tube relationship for TH

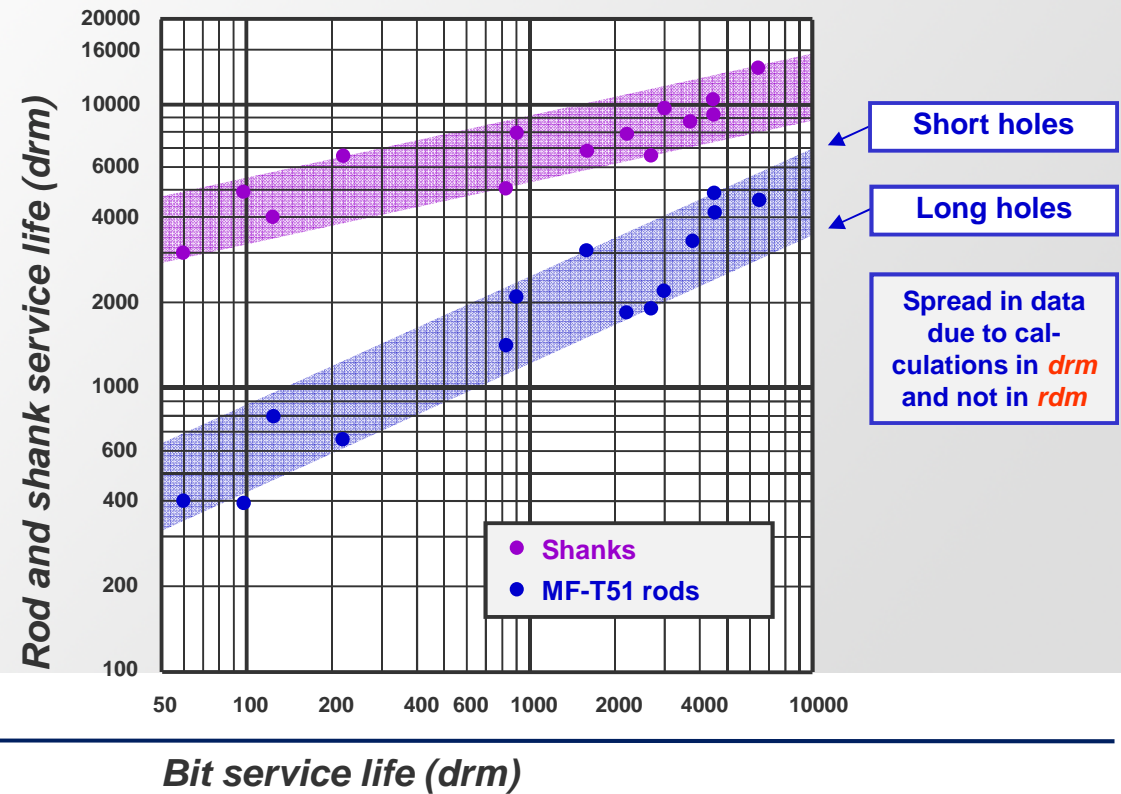


Jobsite KPI's for drill steel

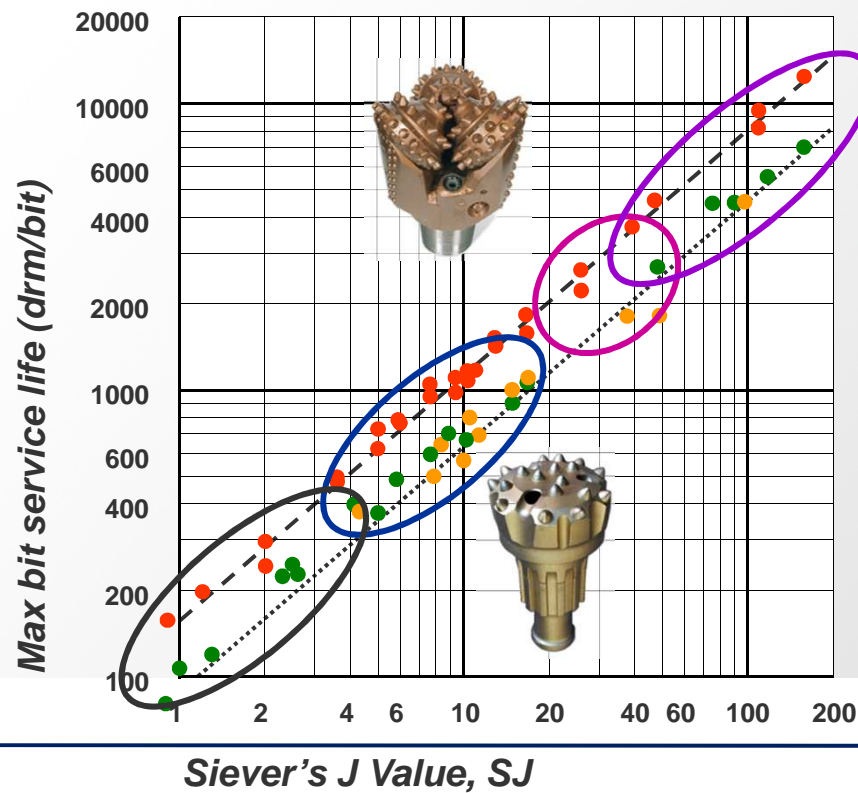
- drill steel component life
- bit regrind intervals
- bit replacement diameter
- component discard analysis
- costs in € per drm or m³



Example of drill steel followup for MF-T51



Trendlines for bit service life



- Rotary Drilling - Ø311mm / Std.
 - DTH *
 - Tophammer *
- * Bit service life highly dependent on regrind intervals – regard curve as a **top limit value**

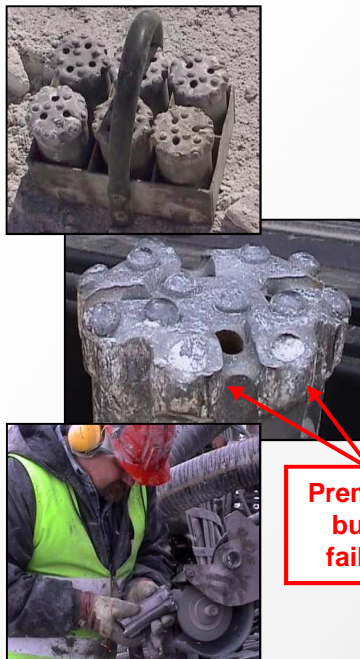
Limestone

Dolomite

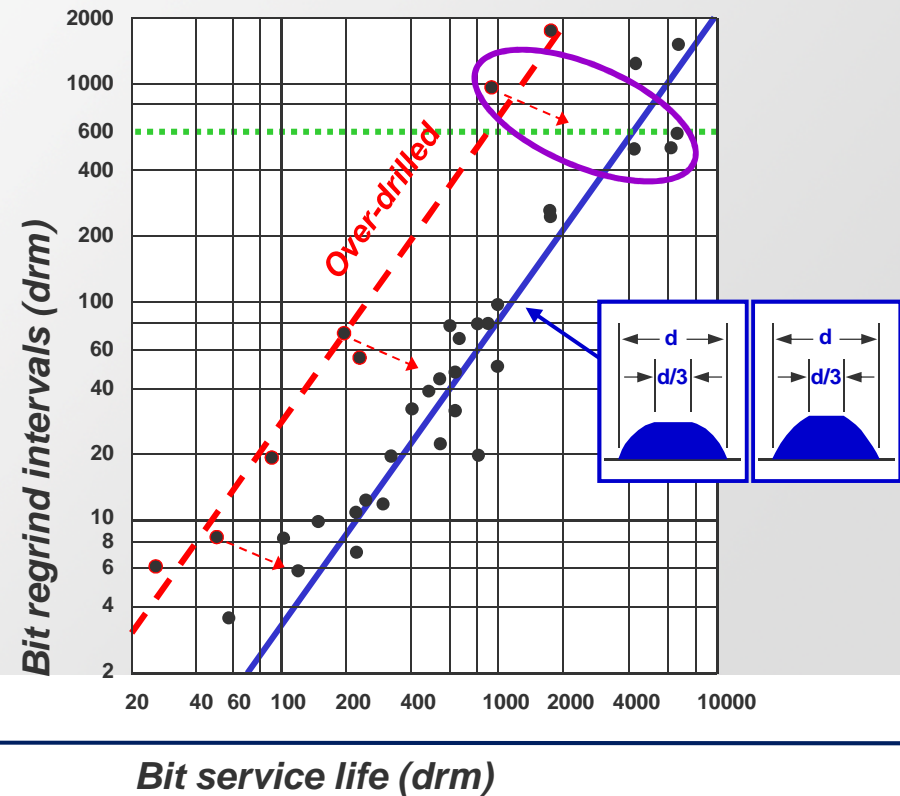
Granite

Quartzite

Bit regrind intervals, bit service life and over-drilling

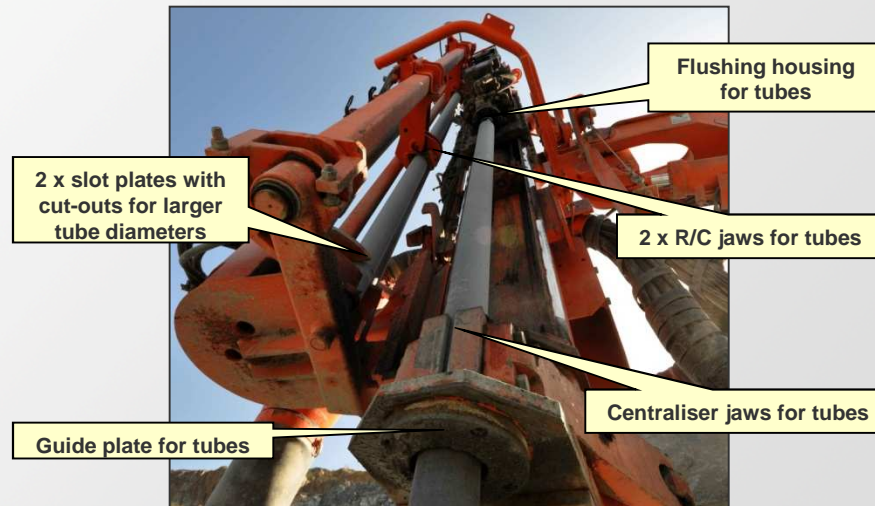


Premature
button
failures

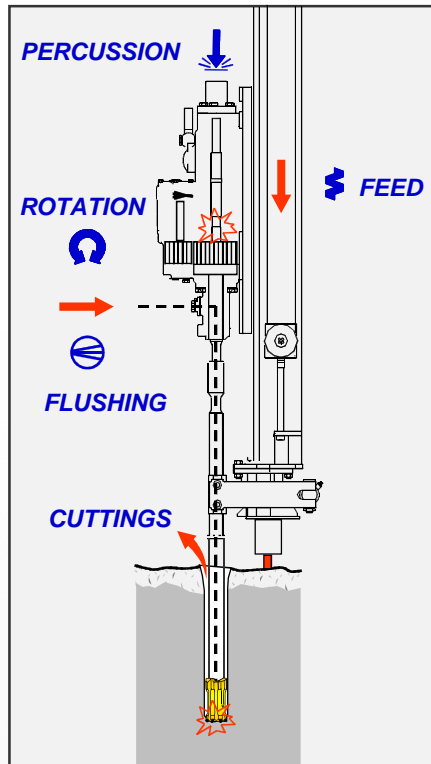


Tube drilling for TH – avoid:

- bending of drill string - leads to premature tube failure
- hardrock drilling or bit service life < 1750 drm - leads to poor tube life
- jerky drill string rotation - leads to an unstable bit which initiates drill-hole deviation



Mechanics of percussive drilling



Percussive drilling

- ✓ **Down-the-hole, DTH**
Stress waves transmitted directly through bit into rock
- ✓ **Tophammer**
Stress wave energy transmitted through shank, rods, bit, and then into rock

Basic functions

- ✓ **percussion** - reciprocating piston used to produce stress waves to power rock indentation
- ✓ **feed** - provide bit-rock contact at impact
- ✓ **rotation** - provide bit impact indexing
- ✓ **flushing** - cuttings removal from hole bottom
- ✓ **foam flushing** - drill-hole wall stabilisation

Flushing of drill-cuttings

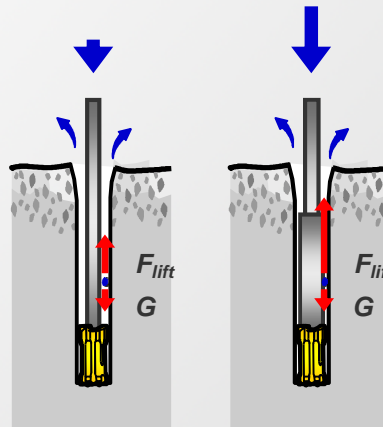
Insufficient air < 15 m/s

- low bit penetration rates
- poor percussion dynamics
- interrupt drilling to clean holes
- plugged bit flushing holes
- stuck drill steel
- "circulating" big chip wear



Too much air > 30 m/s

- excessive drill steel wear
- erosion of hole collaring point
- extra dust emissions
- increased fuel consumption



Correction factors

- high density rock
- badly fractured rock
(air lost in fractures
- use water or foam to
mud up hole walls)
- high altitude
(low air density)
- large chips require
additional air as well

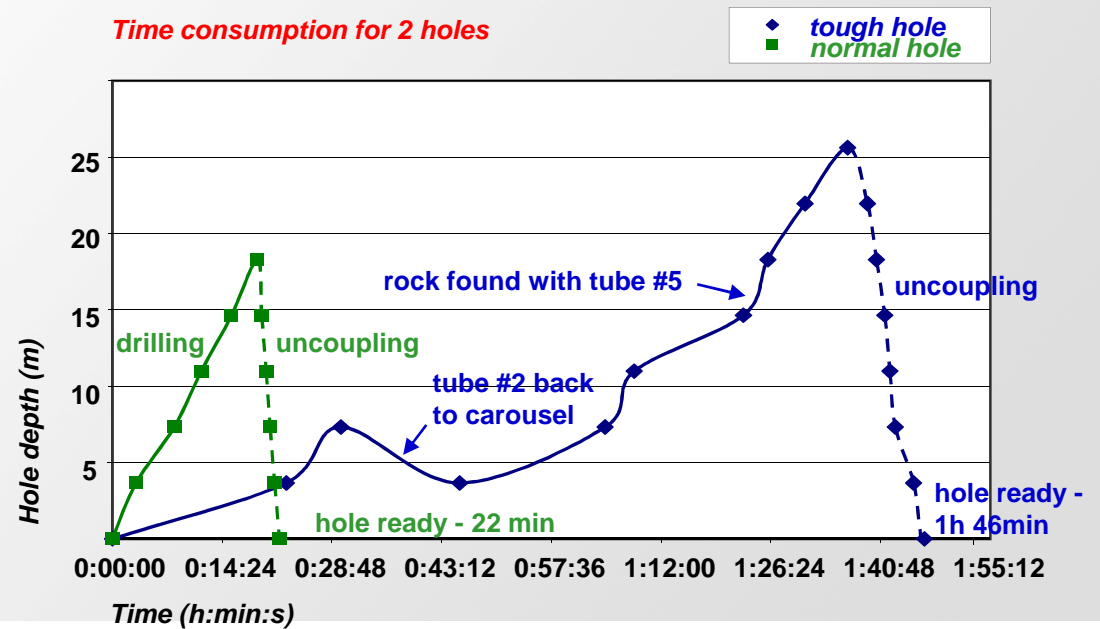


Foam flushing – an aid for drilling in caving material

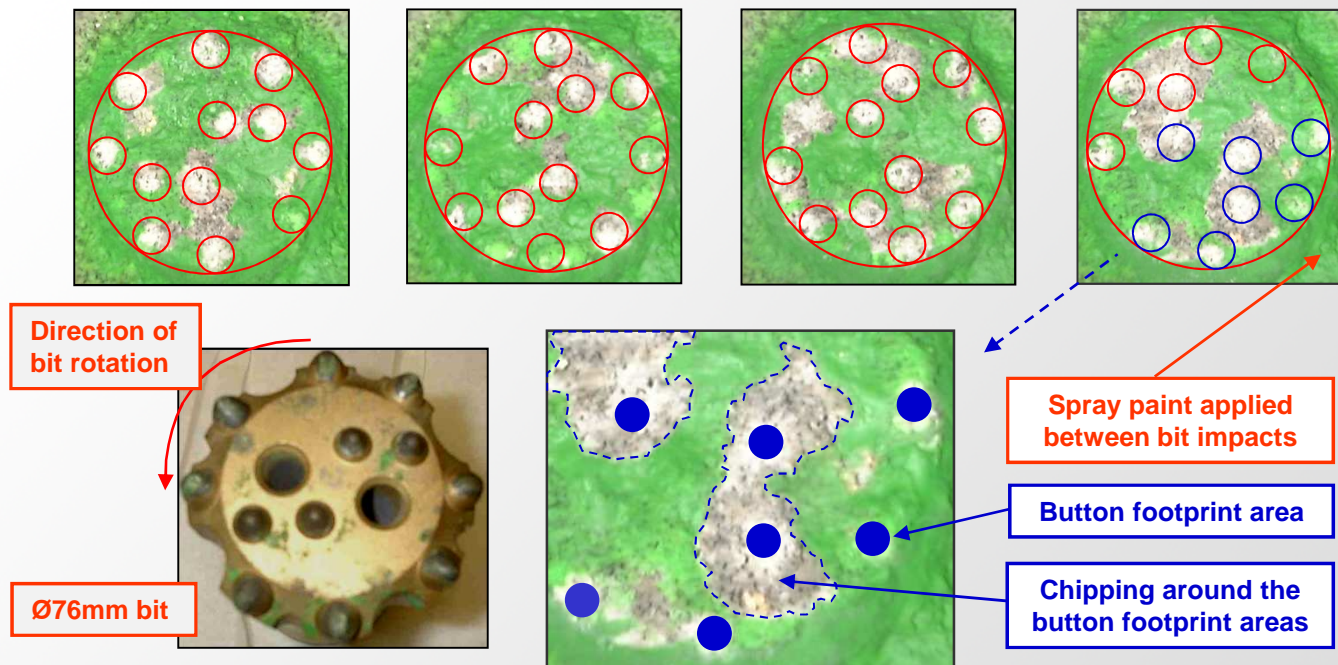


Burst of inhole water

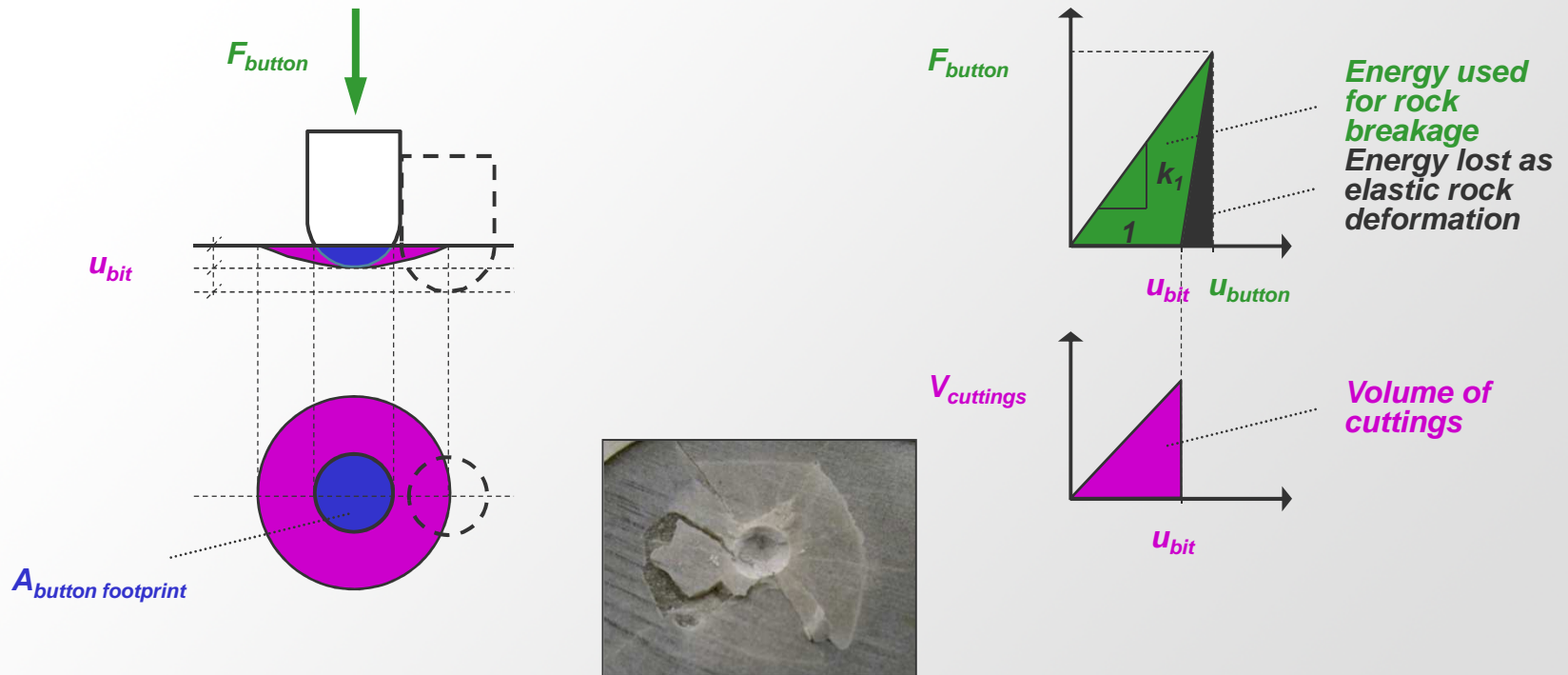
Time consumption for 2 holes



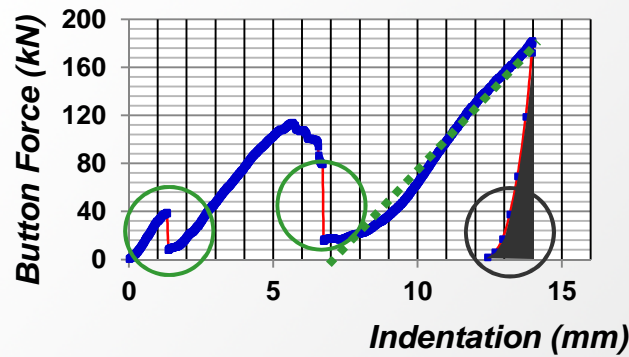
Chip formation by bit indentation and indexing



How rock breaks by indentation



Indentation with multiple chipping

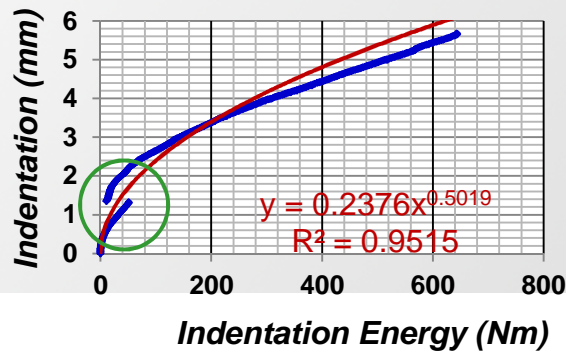


--- $k_1 = 30 \text{ kN/mm}$ for on-loading
— $k_2 = 112.5 \text{ kN/mm}$ for off-loading
 $\gamma = k_1 / k_2 = 0.267$

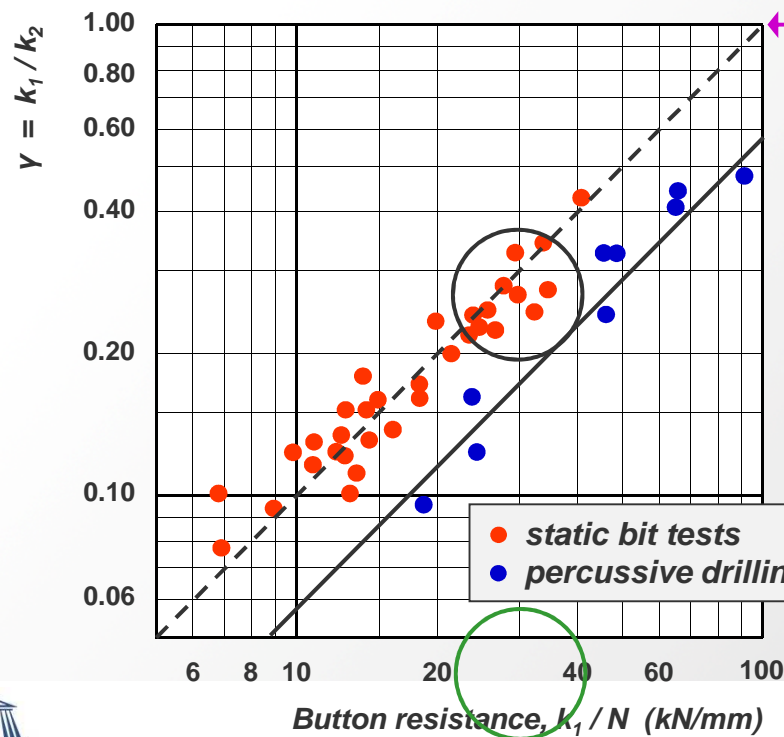
Drop in button force indicates elastically stored energy in rock released by chipping

chipping while on-loading

chipping after off-loading



Energy transfer efficiency η related to rock chipping



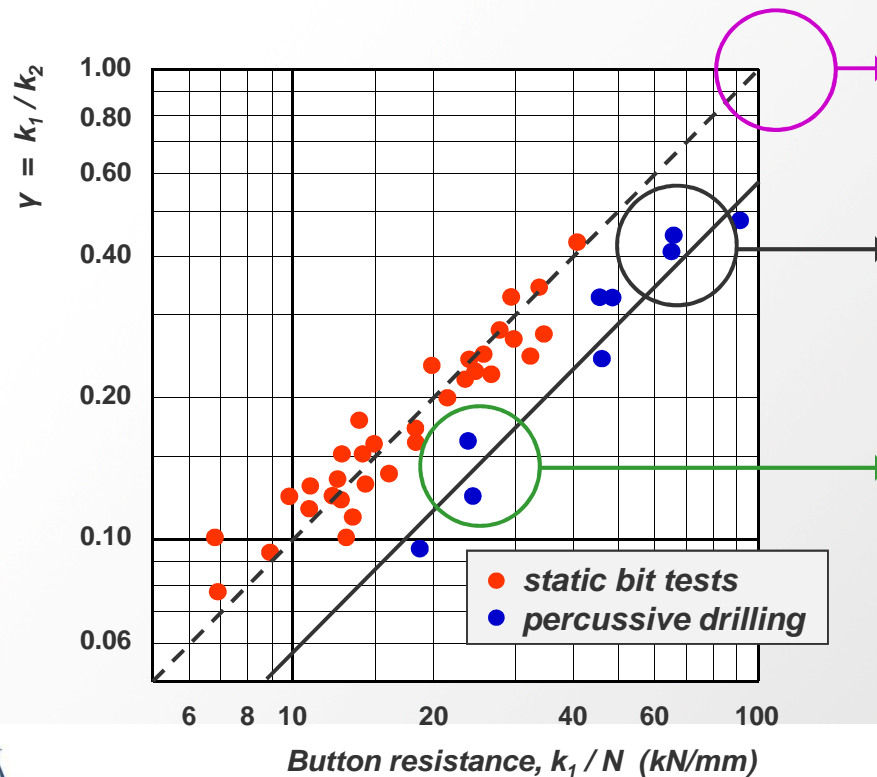
No energy retained in rock after off-loading for $\gamma = 1.0$ (all elastic energy in rock returned to drill string)

$$\eta = W_{rock} / W_{incident}$$

$$= \eta_{impedance} \cdot (1 - \gamma)$$

$$\eta_{impedance-max} \approx 0.90$$

How does this apply to practical drilling?

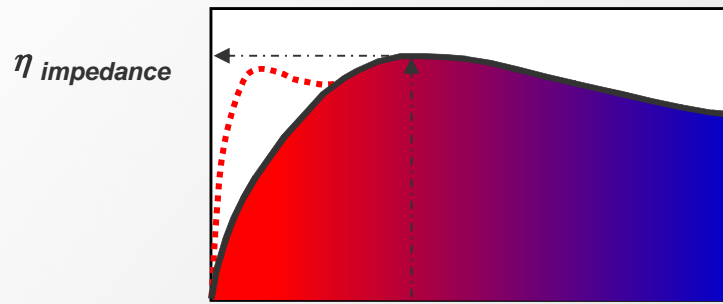
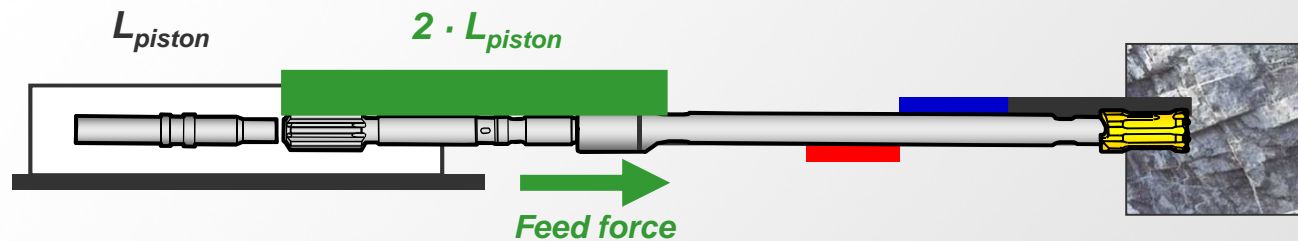


“Hopeless” drilling situation
- no bit advance rate for $\gamma = 1.0$

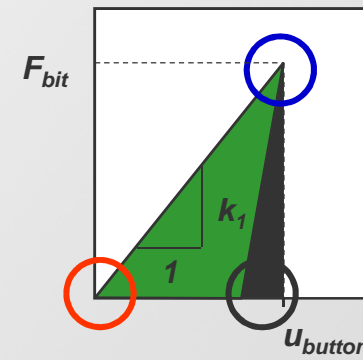
“Poor” drilling situation
- chipping after off-loading
- potential for severely reduced drill steel life
- correct choice of bit design and size?
- sufficient bit regrinding (resharpening)?

“Good” drilling situation
- chipping during on-loading
- potential to achieve max drill steel service life

Energy transfer efficiency η related to impedance matching

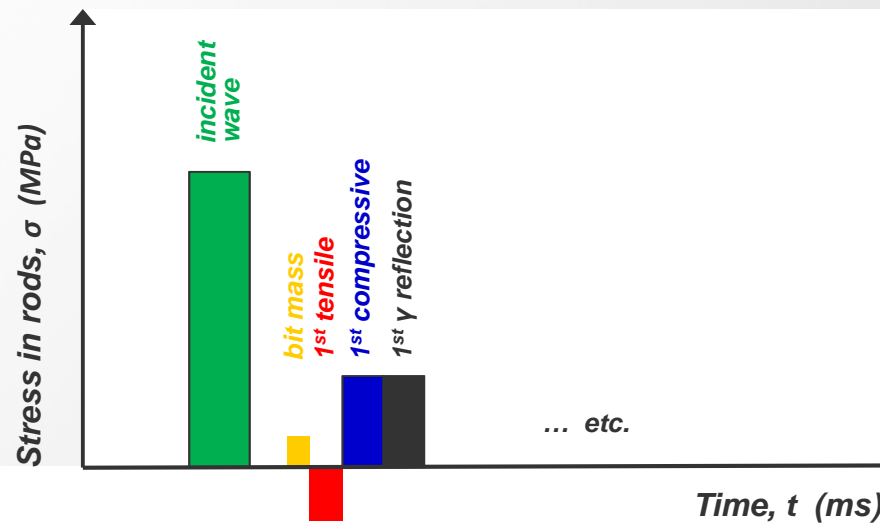


$k_1 = \text{bit indentation resistance (kN/mm)}$

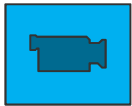


How do we study drill string energy transfer issues?

- strain gauge measurements on rods/tubes while drilling
 - numerical modelling
- => the tell-tale items we are looking for:



Energy transfer chain – video clip cases



cavity



“perfect” bit / rock match



bit / rock gap – i.e. underfeed



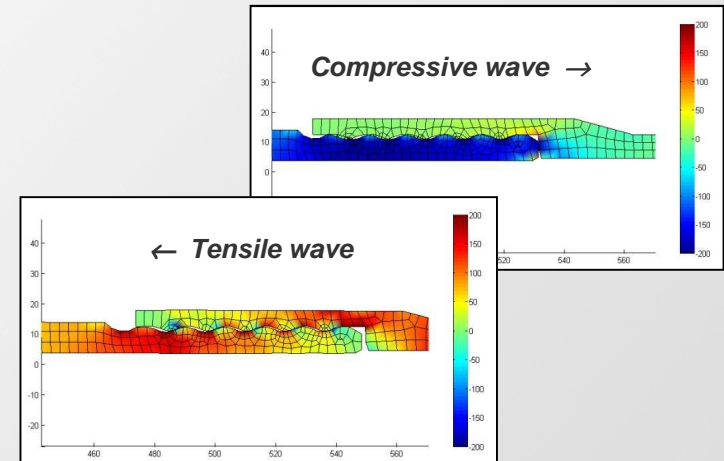
bit face bottoming – caused by:

- *drilling with too high impact energy*
- *drilling with worn bits i.e. too low button protrusion*

Energy transmission through threads

Energy transfer can be divided into:

- **energy transmission through the drill string**
 - optimum when the cross section throughout the drill string is constant
 - length of stress wave
 - weight of bit
- **energy transmission to rock**
 - bit indentation resistance – k_1
 - bit-rock contact



The most critical issue in controlling stress waves is to avoid high tensile reflection waves.

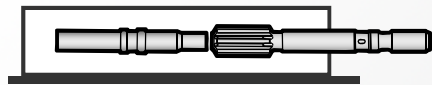
Tensile stresses are transmitted through couplings by the thread surfaces - not through the bottom or shoulder contact as in the case for compressive waves.

High surface stresses combined with micro-sliding result in high coupling temperatures and heavy wear of threads.

Feed force requirements

From a drilling point of view

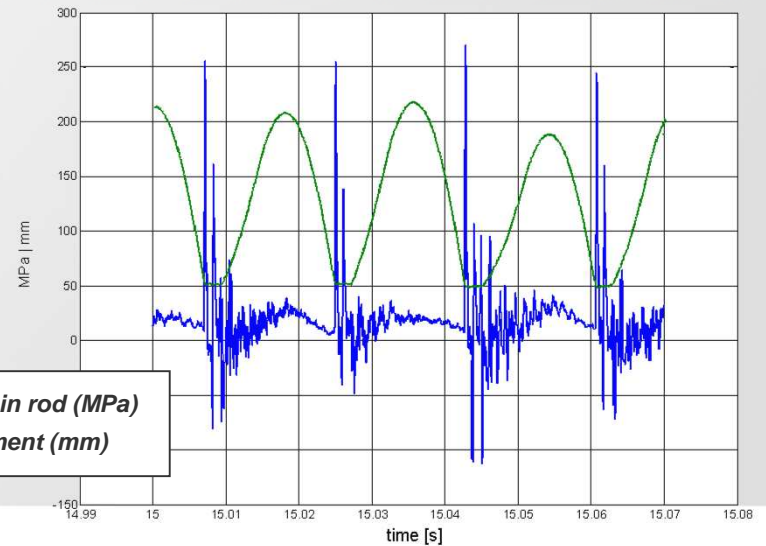
- to provide bit-rock contact
- to provide rotation resistance so as to keep threads tight



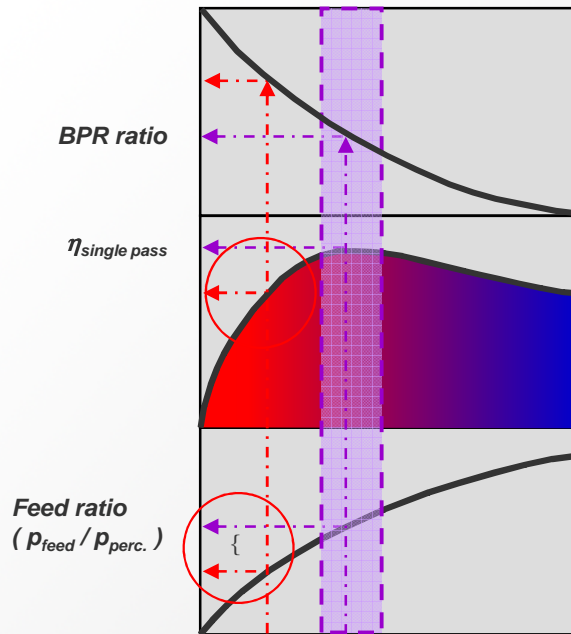
← Rod force
→ Feed force

From a mechanical point of view

- compensate piston motion
- compensate linear momentum of stress waves in rods



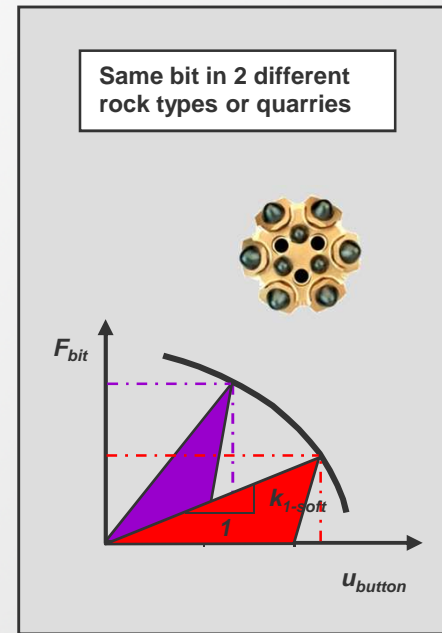
Matching drill settings to site conditions



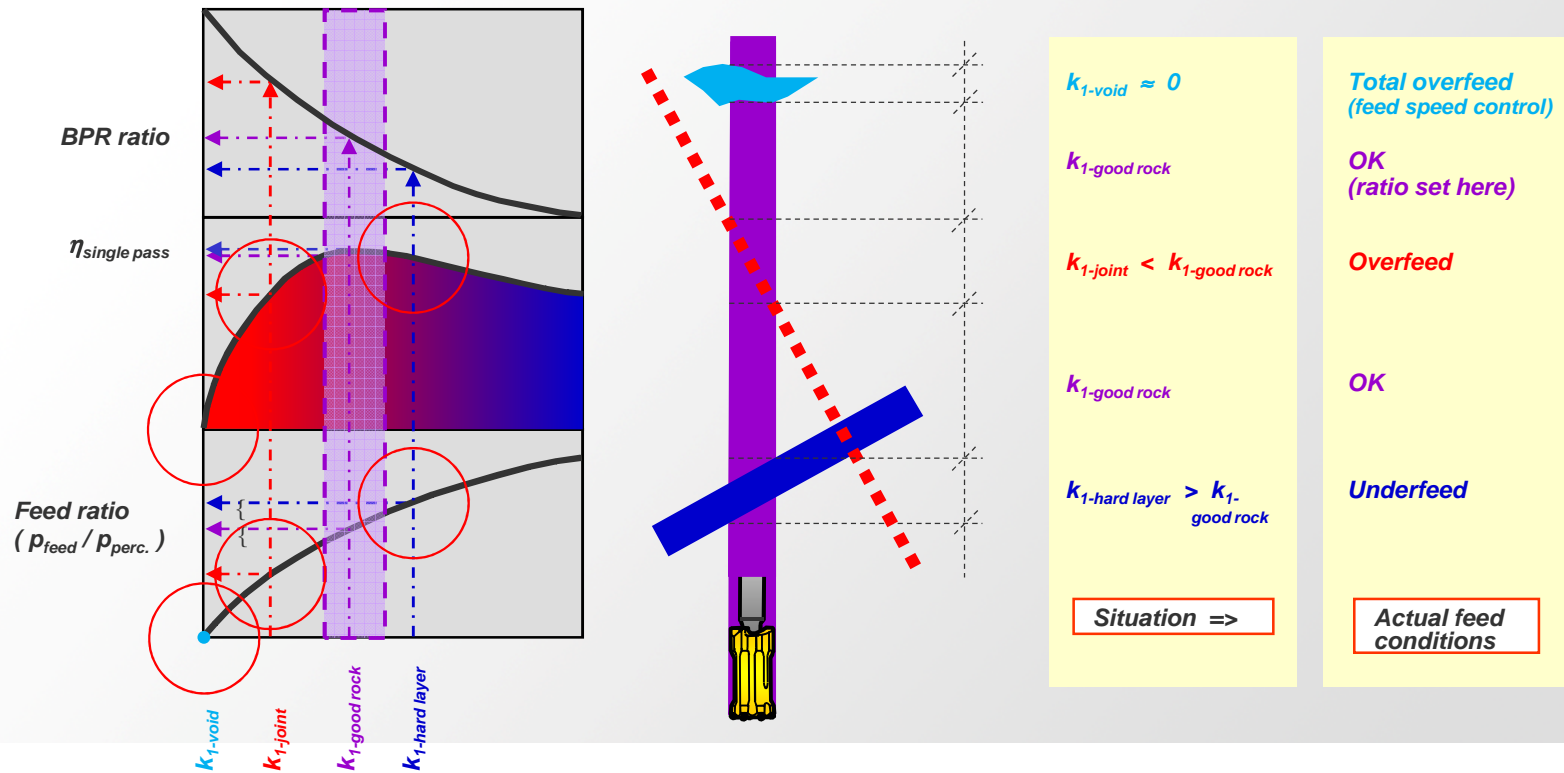
k_{1-soft} $k_{1-good\ rock}$

Rock hardness \dashrightarrow

Button count and size
(and bit size) \dashrightarrow



Drilling in variable rock mass conditions

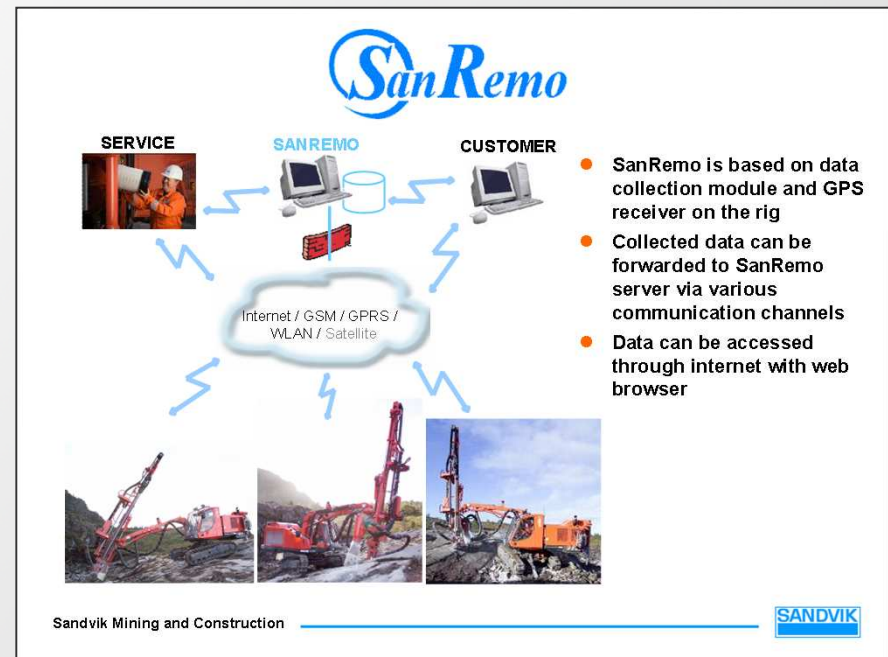


Jobsite KPI's for drilling operations

- drilling capacities in drm/ph or drm/eh
- production capacity in drm/shift
- avg. percussion pressure
- fuel consumption l/eh
- drill steel consumption & costs

- drill-hole straightness
- geological conditions

- costs in € per drm or m³



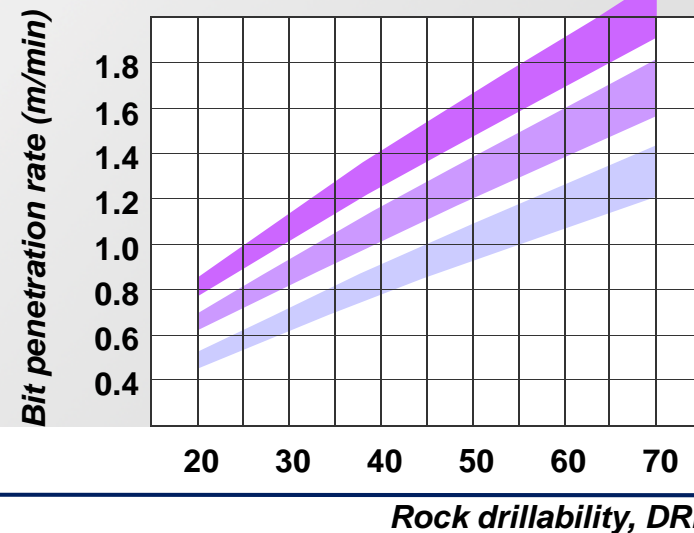
Predicting bit penetration rates - TH

- rock mass drillability, DRI
- percussion power level in rods (tubes)
- bit diameter and type
 - ✓ hole wall confinement of gauge buttons
- goodness of hole-bottom chipping
 - ✓ bit face design and insert types
 - ✓ drilling parameter settings (RPM, feed)
- flushing medium and return flow velocity

HL1060T	102 mm	4"
HF1560T	115 mm	4.5"
HL1010	102 mm	4"

HL650T	76 mm	3"
HL810T	89 mm	3.5"
HF810T	89 mm	3.5"
HL1560T	115 mm	4.5"

HL510/HLX5	76 mm	3"
HL710	89 mm	3.5"



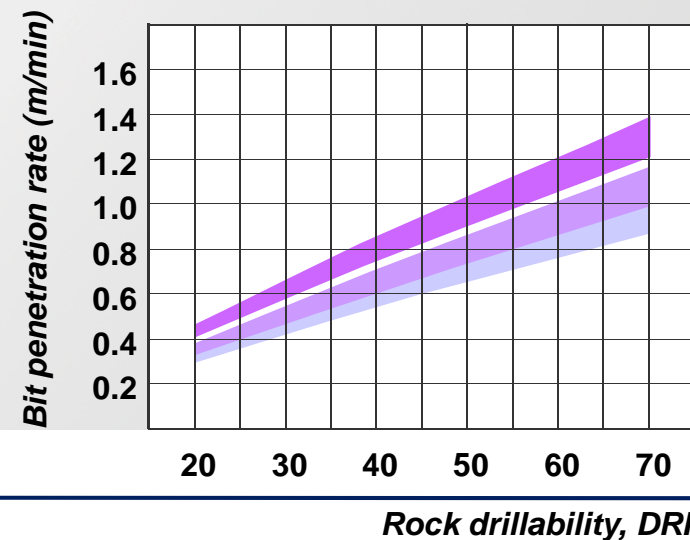
Predicting bit penetration rates - DTH

- *rock mass drillability, DRI*
- *percussion power of hammer*
- *bit diameter and type*
 - ✓ *hole wall confinement of gauge buttons*
- *goodness of hole-bottom chipping*
 - ✓ *bit face design and insert types*
 - ✓ *drilling parameter settings (RPM, feed)*
- *flushing medium and return flow velocity*

5" RH550 (M50)	140 mm	5.5"
6" RH550 (M60)	165 mm	6.5"

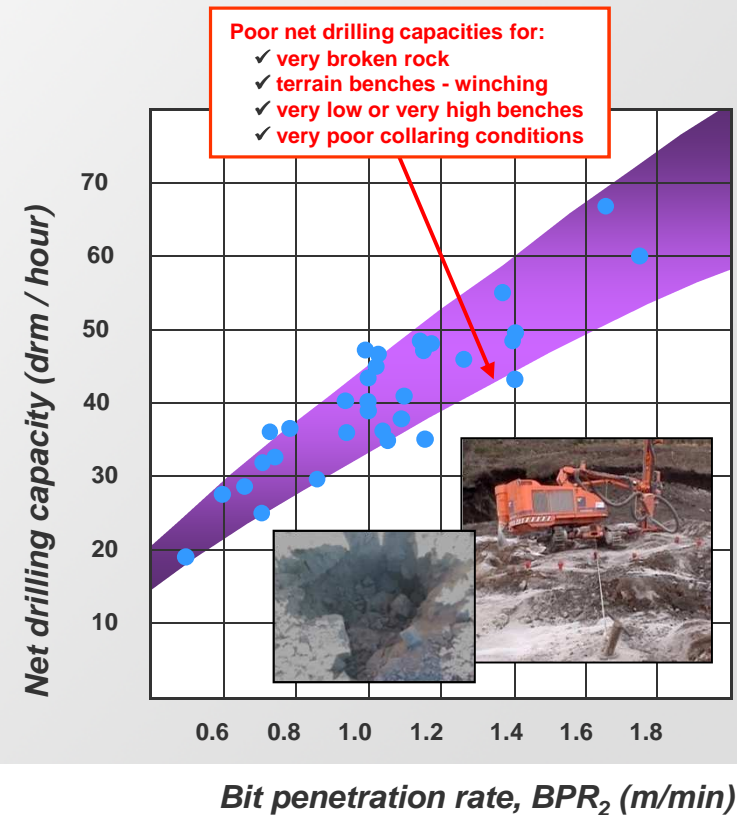
3" RH550 (M30)	89 mm	3.5"
4" RH550 (M40)	115 mm	4.5"
6" RH550 (M60)	203 mm	8"

8" RH550 (M85)	251 mm	7/8"
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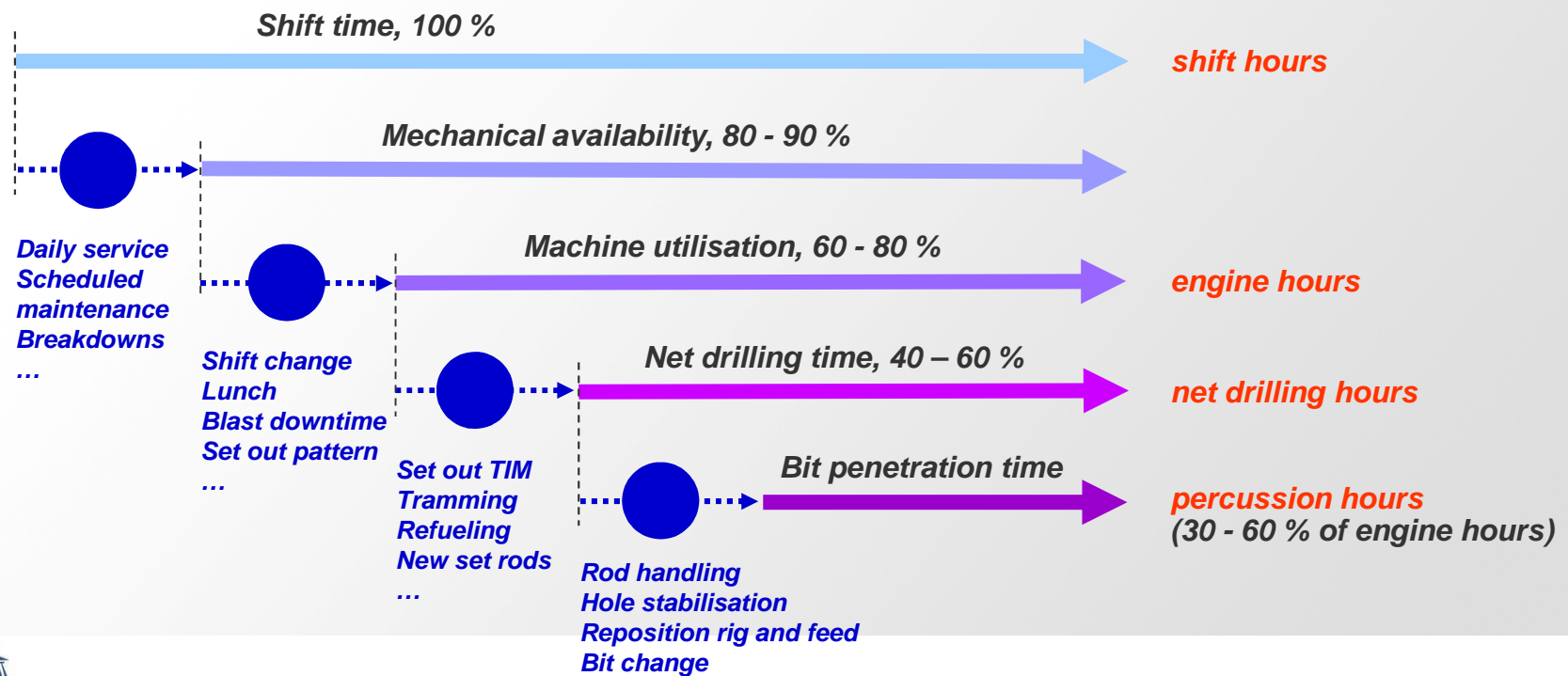


Gross drilling capacities (drm/h)

- *time for rig setup and feed alignment per drill-hole*
- *collaring time through overburden or sub-drill zone*
- *drill-hole wall stabilisation time (if required)*
- *rod handling times (unit time and rod count)*
- *bit penetration loss rate percentage i.e.*
 - ✓ *rods and couplings* **6.1 % per rod**
 - ✓ *MF rods* **3.6 % per rod**
 - ✓ *tubes* **2.6 % per tube**
- *effect of percussion power levels on:*
 - ✓ *bit penetration rates*
 - ✓ *drill steel service life*
 - ✓ *drill-hole straightness*
- *time for tramming between benches, refueling, etc.*
- *effect of operator work environment on effective work hours per shift*
- *rig availability, service availability, service and maintenance intervals*



Typical breakdown of longterm rig usage and capacities



Limestone Quarry – Drilling Report DI550 / Ø140mm

Company: xxxx		Drillmaster: xxxx	
1. Net drilling time	43,0 --->	39,5	Net drilling time (drm/ph) - percussion hours
2. Moving :	10,7		
3. Total 1+2	53,7 --->	31,6	Gross drilling time (drm/h) - incl. time for moving on bench
4. No-productive time	6,1h	Waiting: h	
		Re-dress: h	Maintenance: 4,1h
		Re-fuel: h	Repair: 2,0h
		Total: h	Total: 6,1h
5. Driving time	4,3h		
6. Shift time 3-5 (without breaks)	70,50 --->	24,1	drm/shift hour
	Efficiency (NG) :	76%	Availability: 89%
	Oper. ratio (eh/sh) :	80%	
7. Fuel consumption			
Total consumption:	3131,0 L --->	1,84	l/drm meter
	--->	53,98	l/engine hour

Can we drill straight holes?

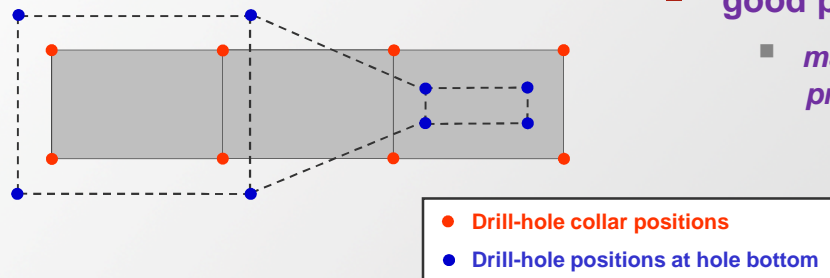
Ventilation Shaft, Olkiluoto Nuclear Power Plant

Shaft diameter, Section I	Ø6.5 m
Shaft length	15 m
Rock type	Quartz Diorite
Contour hole size	Ø60 mm
Contour hole charging	80 g/m det. cord
Contour hole spacing	0.4 m
Contour row burden	0.7 m



What happens when we shoot holes that look like spaghetti?

- floor humps
 - *poor loading conditions, uneven floors*
- poor walls
 - *unstable walls*
 - *difficult 1st row drilling*
- flyrock
 - *safety issue*
- blowout of stemming
 - *safety, dust, toes, ...*
- blast direction
 - *quality of floors and walls*
- shothole deflagration / misfires
 - *safety*
 - *locally choked muckpiles (poor diggability)*
- good practise
 - *max. drill-hole deviation up to 2 – 3 % for production drilling*



How do we go about drilling straighter holes?

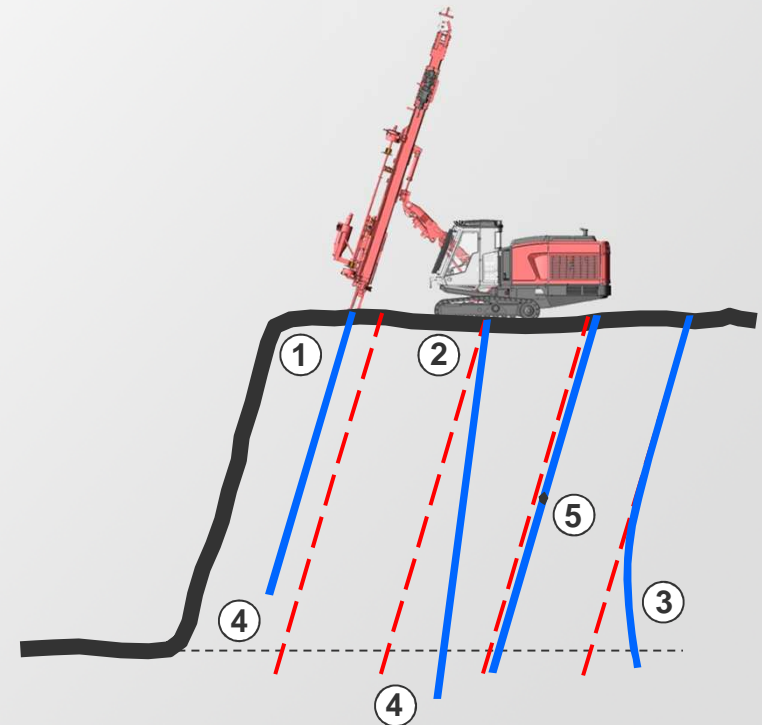
- understand the many issues leading to drill-hole deviation
- technically good drill string
- technically good drill rig, instrumentation, ...
- **motivate the drillers!**



Accurate drilling gives effective blasting

Sources of drilling error

1. Collar position
2. Hole inclination and direction
3. Deflection (bending)
4. Hole depth
5. Omitted or lost holes
6. Shothole diameter (worn out bits)



Examples of drill-hole deviation

Directional error for
Ø89 mm retrac bit /
T45 in granite



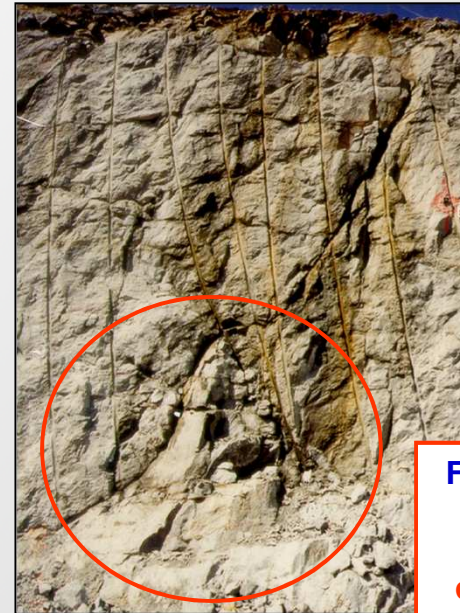
Drill string deflection caused
by gravitational pull or
sagging of drill steel in
inclined holes in syenite



Examples of drill-hole deviation



**Deflection with and without pilot tube for
Ø89 mm DC retrac bit / T51 in micaschist**



**Floor hump due
to explosives
malfunction -
caused by drill
string deflection**

Shothole diameter error control

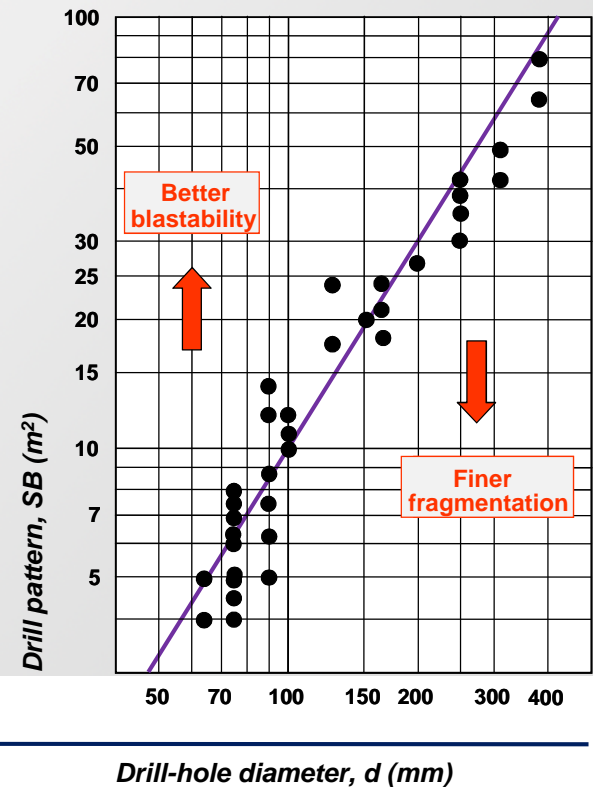
- bits loose diameter due to gauge button wear
- typical diameter loss for worn out bits is ~ 10%
- diameter loss effect on drill patterns

Diameter new bit $\varnothing 102\text{mm}$

Diameter worn out bit $\varnothing 89\text{mm}$

Diameter loss $(102 - 89) / 102 = 12.8\%$

\Rightarrow *Drill pattern too big* $(102 / 89)^{1.6} = 24\%$



Collar position error control

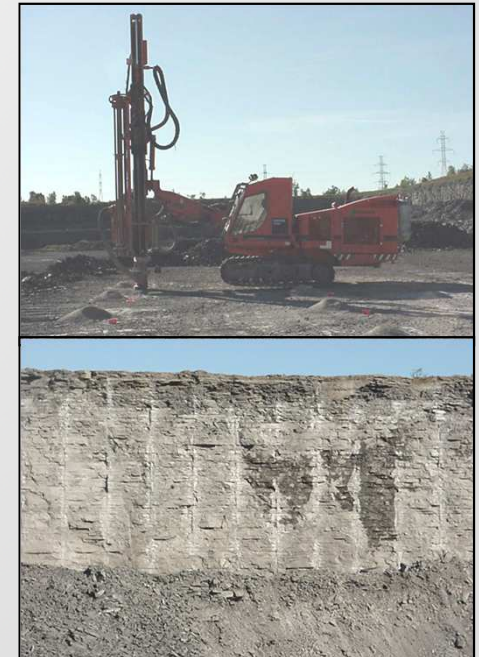
- use tape, optical squares or alignment lasers for measuring in collar positions
- use GPS or total stations to measure in collar positions
- collar positions should be marked using painted lines – not movable objects such as rocks etc.
- completed drillholes should be protected by shothole plugs etc. to prevent holes from caving in (and filling up)
- *use GPS guided collar positioning devices e.g. TIM-3D*



Difficult 1st row drilling

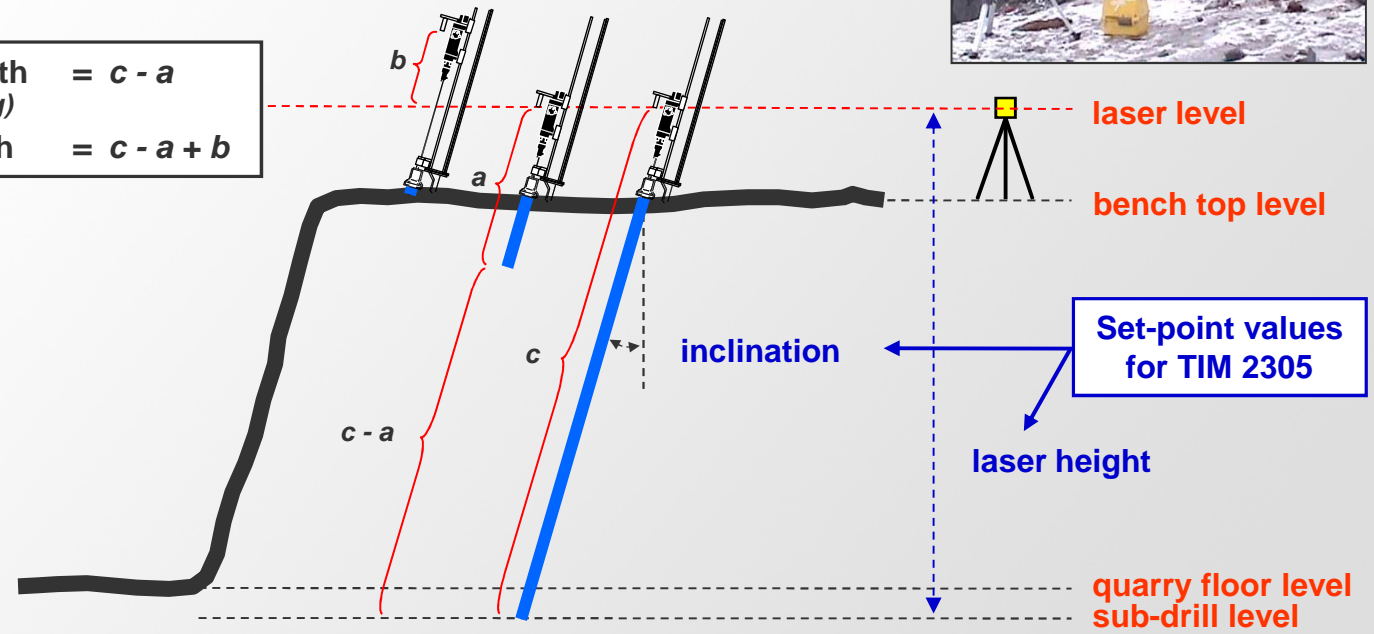
Lafarge Bath Operations, Ontario

■ Rock type	limestone, 1.6 Mtpa
■ Bench height	32 m
■ Bit	Ø115 mm guide XDC
■ Drill steel	Sandvik 60 + pilot tube
■ Hole-bottom deflection	< 1.5 % or 0.5 m
■ Gross drilling capacity	67 drm/h
■ Drill pattern	4.5 x 4.8 m ² (staggered)
■ Sub-drill	0 m (blasted to fault line)
■ Stemming	2.8 m
■ No. of decks	3 (stem between decks 1.8 m)
■ Deck delays	25 milliseconds
■ Charge per shothole	236 kg
■ Explosives	ANFO (0.95 & 0.85 g/cm ³)
■ Powder factor	0.34 kg/bm ³

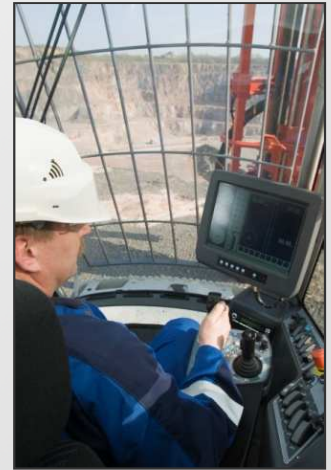
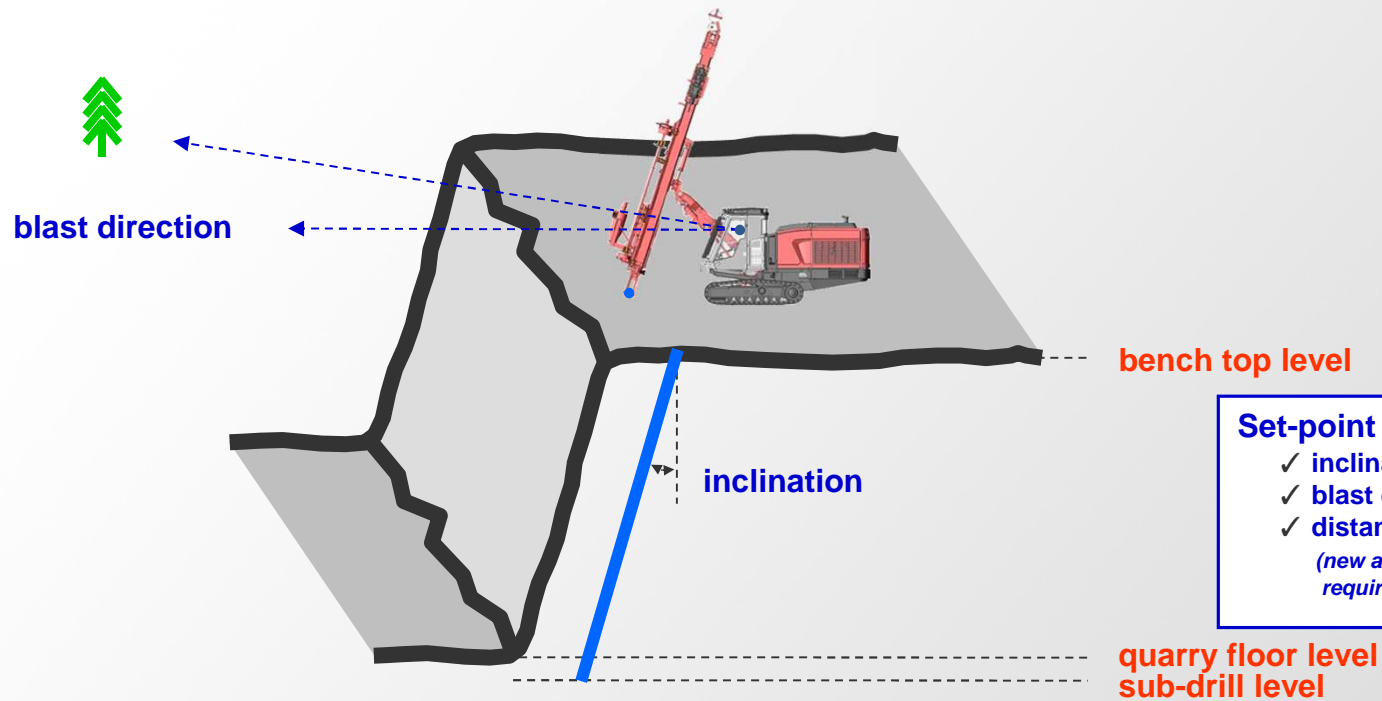


Hole depth error control

Remaining drill length = $c - a$
(at 1st laser level reading)
Total drill hole length = $c - a + b$

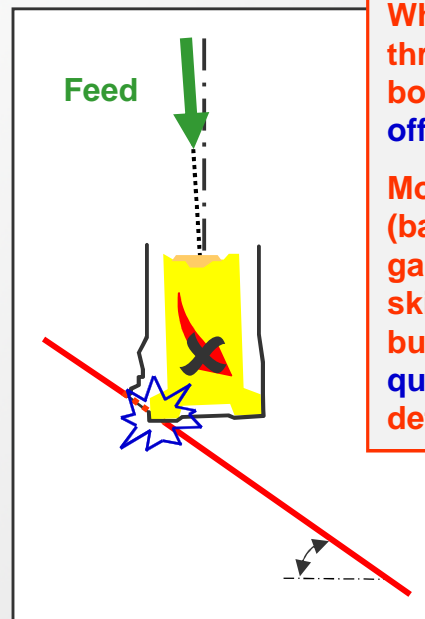


Inclination and directional error control



- Set-point values for TIM 2305**
- ✓ inclination
 - ✓ blast direction projection
 - ✓ distant aiming point direction
- (new aiming point reading required when tracks are moved)*

How bit face designs enhance drill-hole straightness

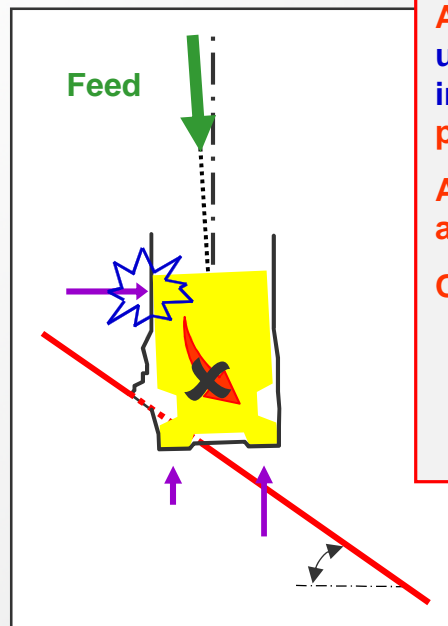


When the bit first starts to penetrate through the joint surface on the hole bottom - the gauge buttons tend to skid off this surface and thus deflect the bit.

More aggressively shaped gauge inserts (ballistic / chisel inserts) and bit face gauge profiles (drop center) reduce this skidding effect by enabling the gauge buttons to “cut” through joint surfaces quickly - thus resulting in less overall bit deflection.



How bit skirt designs enhance drill-hole straightness

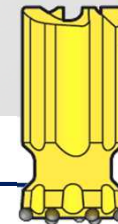


As the bit cuts through the joint surface - an uneven bit face loading condition arises; resulting in bit and drill string axial rotation - which is proportional to bit impact force imbalance.

A rear bit skirt support (retrac type bits) reduces bit and string axial rotation by “centralizing” the bit.

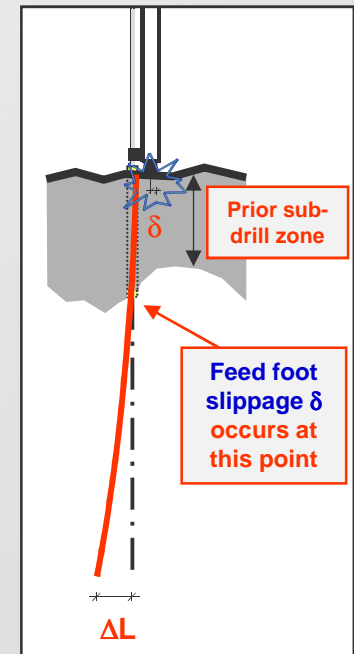
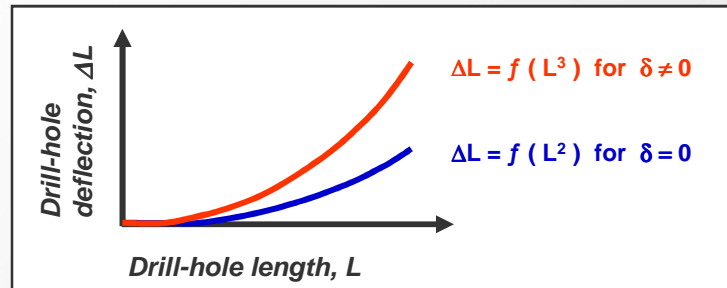
Other counter measures:

- longer bit body
- add pilot tube behind bit
- lower impact energy
- rapid drilling control system reacting to varying torque and feed conditions

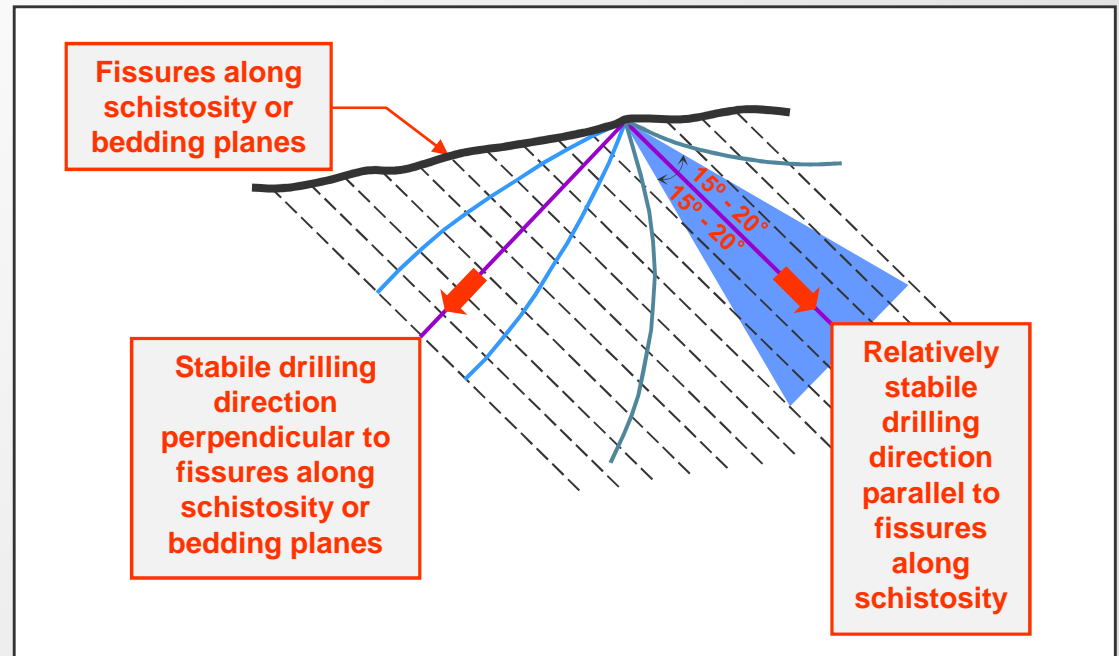


Drill-hole deflection error control

- select **bits** less influenced by rock mass discontinuities
- reduce drill string deflection by using **guide tubes, etc.**
- reduce drill string bending by using less **feed force**
- reduce **feed foot slippage** while drilling - since this causes a misalignment of the feed leading to excessive drill string bending
- avoid **gravitational** effects which lead to **drill string sagging** when drilling inclined shot-holes ($> 15^\circ$)
- avoid inpit operations with **excessive bench heights**

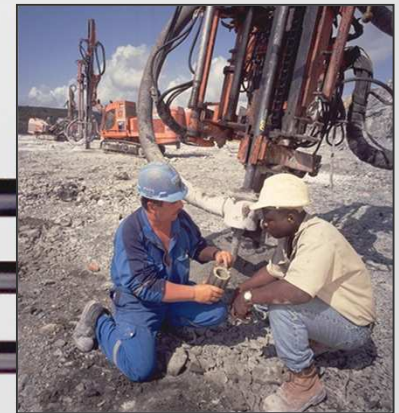
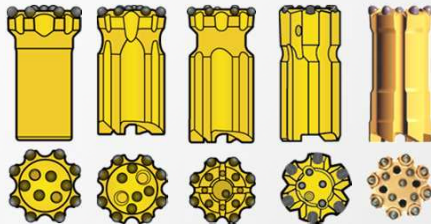


Drill-hole deflection trendlines in schistose rock

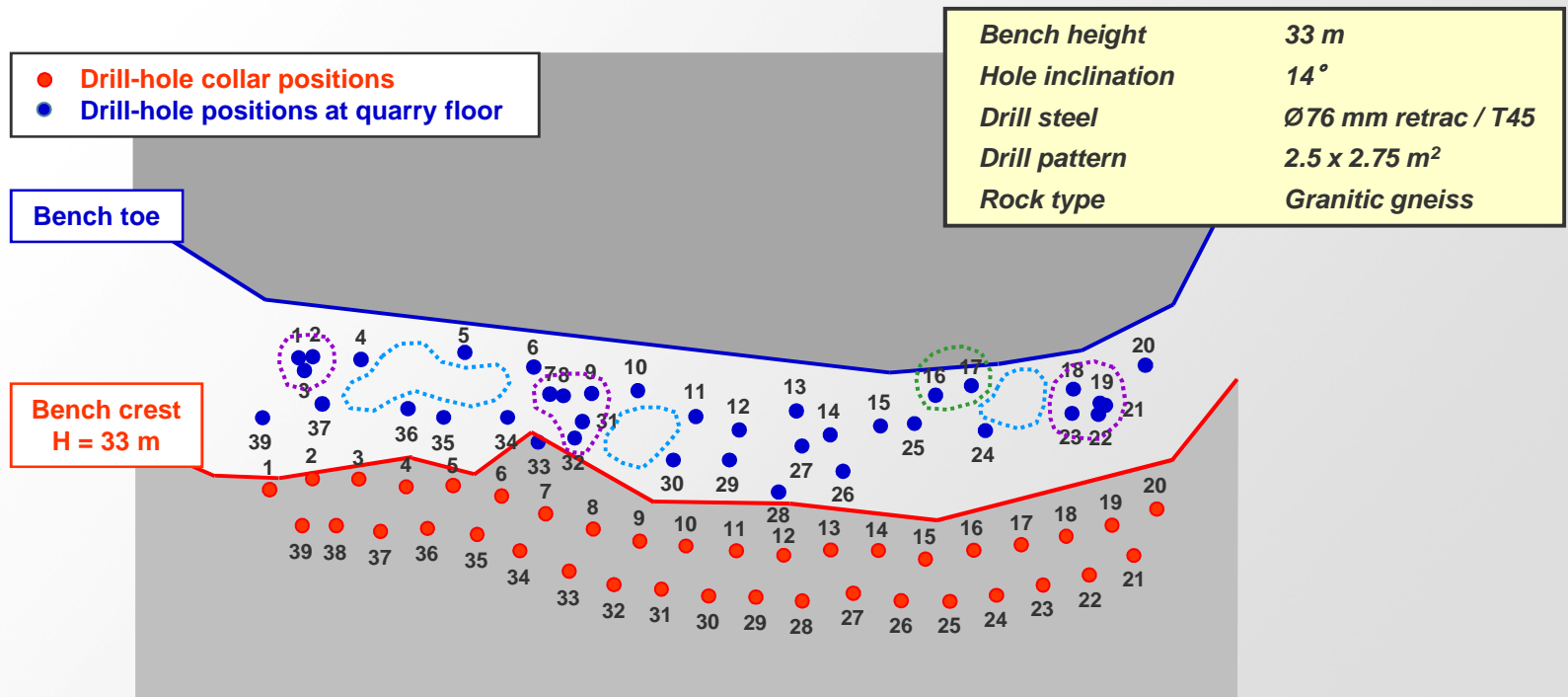


Selecting straight-hole drilling tools - TH

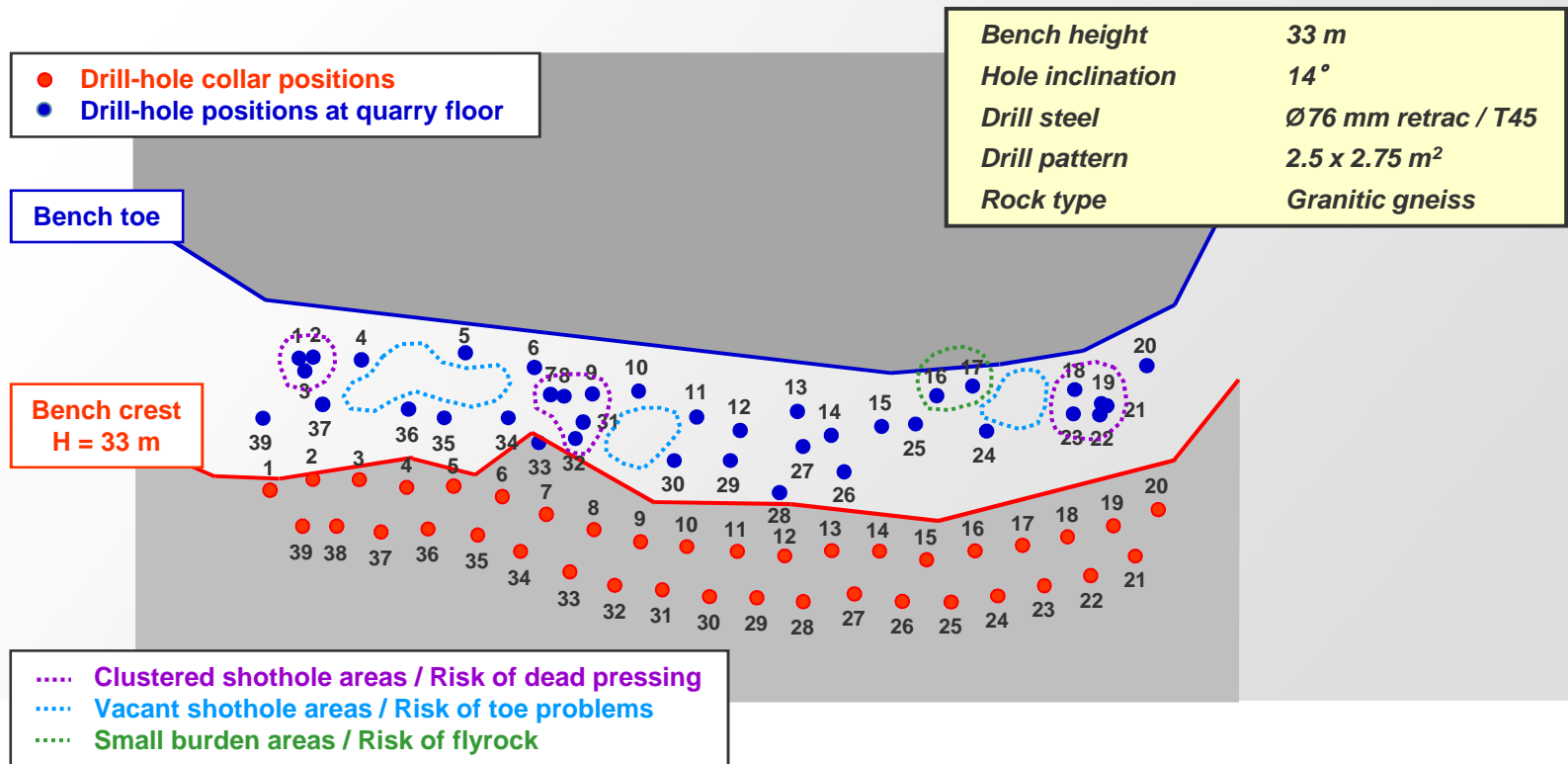
- optimum bit / rod diameter relationship
- insert types / bit face and skirt
 - ✓ spherical / ballistic / chisel inserts
 - ✓ normal bits
 - ✓ retrac bits
 - ✓ drop center bits
 - ✓ guide bits
- additional drill string components
 - ✓ guide tubes / pilot (lead) tubes



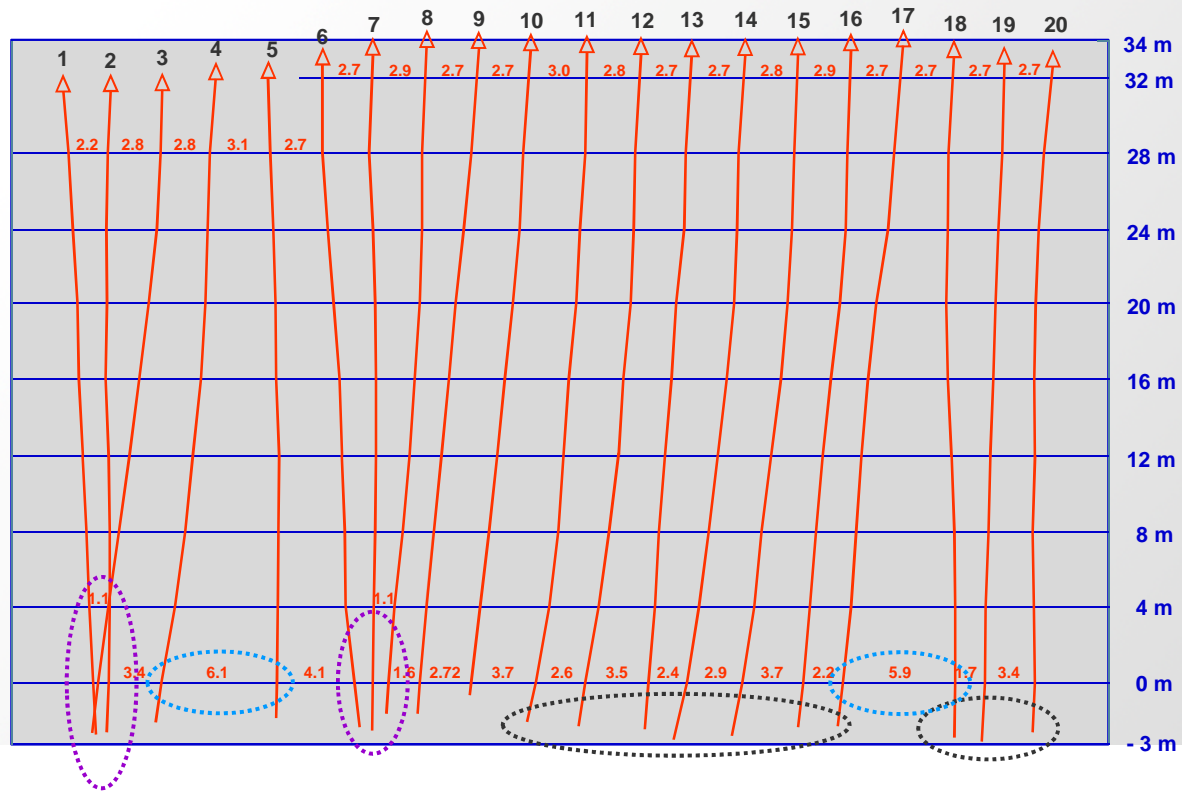
Drill pattern at quarry floor



Drill pattern at quarry floor



Vertical projection of Row 1



Prediction of drill-hole deviation errors

- direction of deviation can not be “predicted”
- magnitude of deviation can be predicted

Rock mass factor, k_{rock}	
■ massive rock mass	0.33
■ moderately fractured	1.0
■ fractured	2.0
■ mixed strata conditions	3.0
Bit design and button factor, k_{bit}	
■ normal bits & sph. buttons	1.0
■ normal bits & ball. buttons	0.70
■ normal X-bits	0.70
■ retrac bits & sph. button	0.88
■ retrac bits & ball. buttons	0.62
■ retrac X-bits	0.62
■ guide bits	0.38

Drill-hole Deviation Prediction				
<i>predH=33.xls/A. Listerud</i>				
Location	Bench H = 33m			
Rock type	Granitic gneiss			
Bit type	Retrac bit			
Bit diameter (mm)	dbit	76		
Rod diameter (mm)	dstring	45		
Guide tube diameter (mm)	dguide / No	No		
Total deflection factor				
rock mass	kdef	1,34		
drill-string stiffness	krock	1,30		
bit wobbling	kstiffness	0,138		
guide tubes for rods	kwobbling	0,592		
bit design and button factor	kguide	1,000		
constant	kbit	0,88		
	krod	0,096		
Inclination and direction error factor				
	k I + D	47,8		
Drill-hole deviation prediction				
Drill-hole Length	Drill-hole Inc + Dir	Drill-hole Deflection	Drill-hole Deviation	Drill-hole Deviation
L	ΔL_{I+D}	ΔL_{def}	ΔL_{total}	$\Delta L_{total} / L$
(m)	(mm)	(mm)	(mm)	(%)
9,3	444	116	459	4,9
13,4	640	241	684	5,1
17,6	840	415	937	5,3
21,7	1036	631	1213	5,6
34,1	1628	1559	2254	6,6

Factors affecting drill-hole deviation

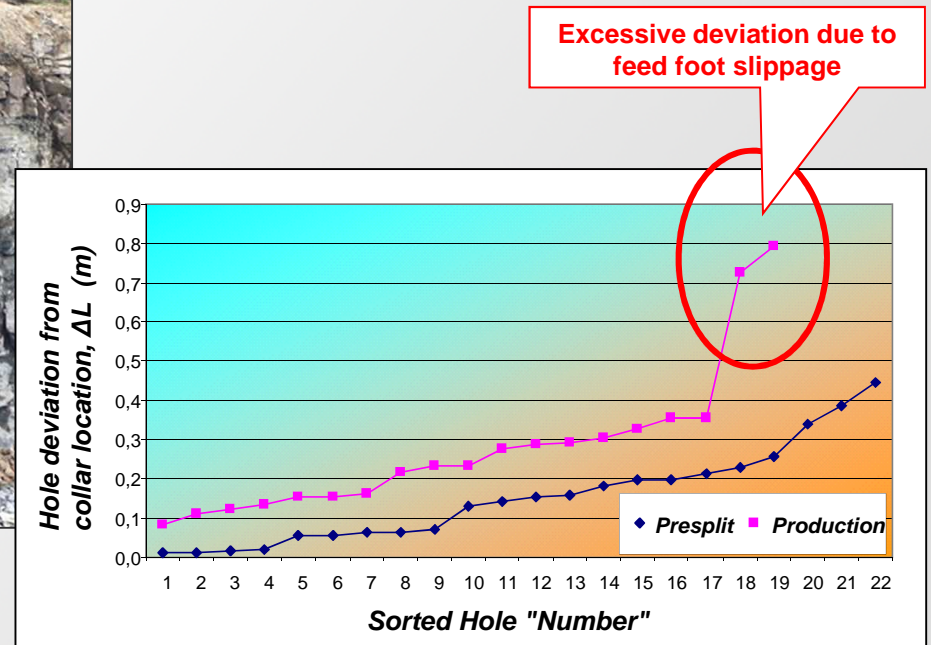
- drill string startup alignment
- bit will follow a joint if at sharp angle to bit path
- drill string stiffness and “tube” steering behind bit
- deviation increases with impact energy
- button shape, bit face and bit body design
- drilling with dull buttons (worn bits)
- bit diameter checks when regrinding
- feed foot slippage while drilling
- removal or controlled drilling through prior sub-drill zone
- drilling control systems, i.e.
 - applied feed, torque and percussion dynamics
- operator motivation!



Wall control drilling Macon Quarry, GA

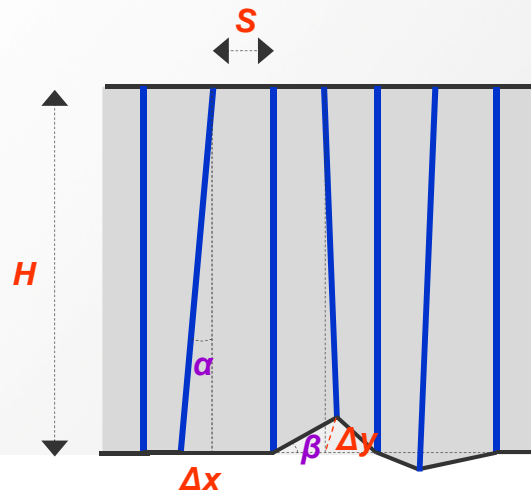


DP1500 - Ø87mm Tubes - 24.5m
Bench - Ø140mm Presplit



Wall control drilling Macon Quarry, GA

	ΔL	%	$\Delta x = \Delta y$	α	β
Max error	0.45 m	1.8	0.32 m ($\approx 2d$)	0.75°	12.0°
Median error	0.15 m	0.6	0.11 m ($\approx d$)	0.25°	4.2°



$$\Delta L = (\Delta x^2 + \Delta y^2)^{1/2}$$

Δy = error perpendicular to wall is of greater importance to extent of blast damage in backwalls

$$\beta = \text{atan } \Delta y / S$$

Δx = error parallel to wall - lesser importance

$$\alpha = \text{atan } \Delta x / H$$

Wall control D&B Chadormalu Iron Mine

Case Study #10 – Presplitting
Chadormalu Iron Mine

	PS	B ₁	B ₂	B ₃	... P
Diameter, mm	195	195	195	251	
Drill Pattern B x S, m ²	5.5 x 1.45	3 x 5.2	4.5 x 5.2		
Charge, kg	18	60	200		
	800mm PVC AZAS (ANFO + TBM)	ANFO	ANFO		

Sandvik Mining and Construction

Case Study #10 – Presplitting
Chadormalu Iron Mine – before/after wall control blasting

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