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# **Energy Release and Failure Model of Coal Samples**

**-Laboratory Test and Numerical Modelling**

**Dr. Ting Ren**

Xiaohan Yang & Lihai Tan

**University of Wollongong**



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AUSTRALIA**



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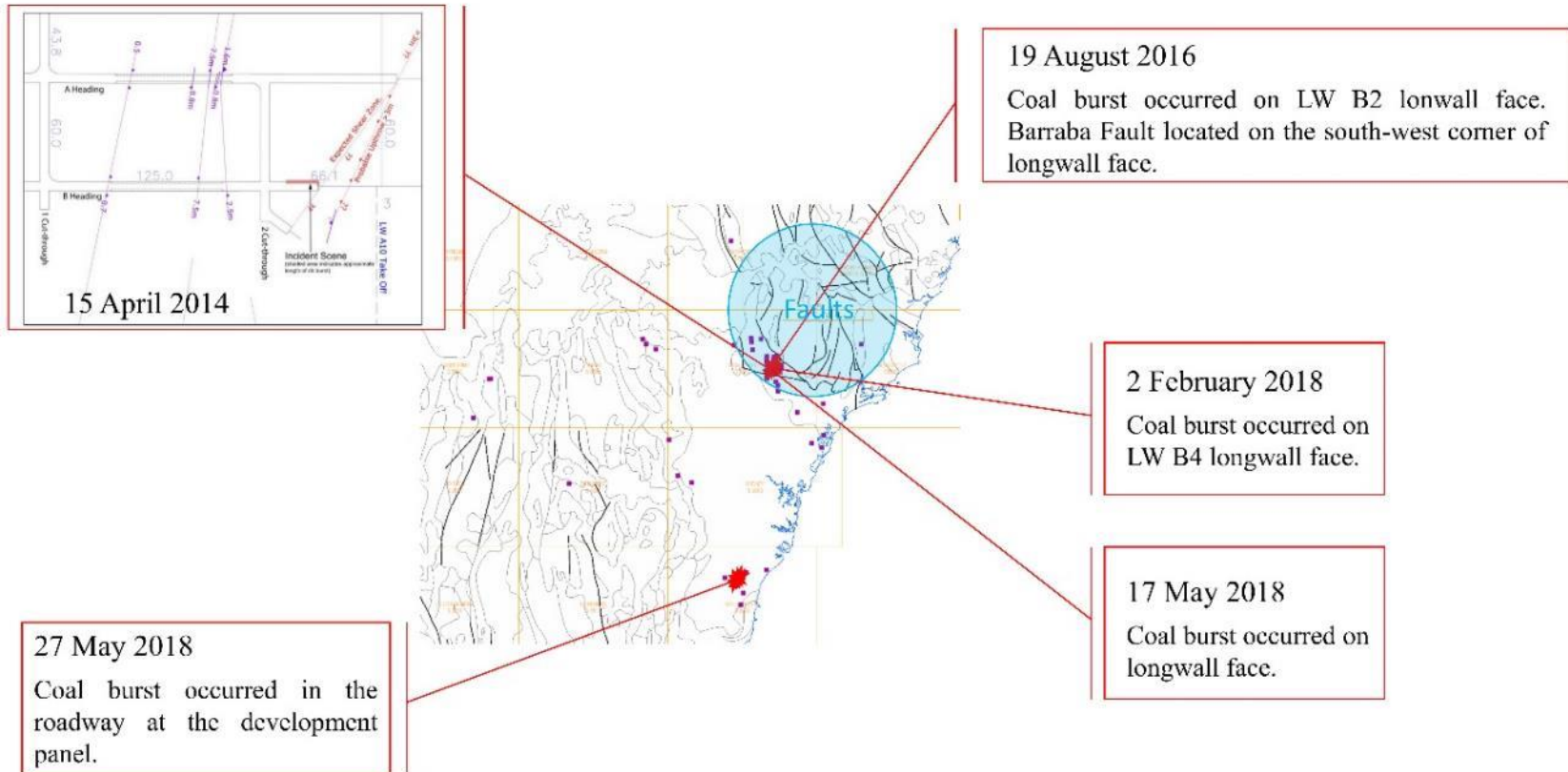
**4. Current Work**

**5. Conclusions**



# Introduction

## Coal Bursts in Australia

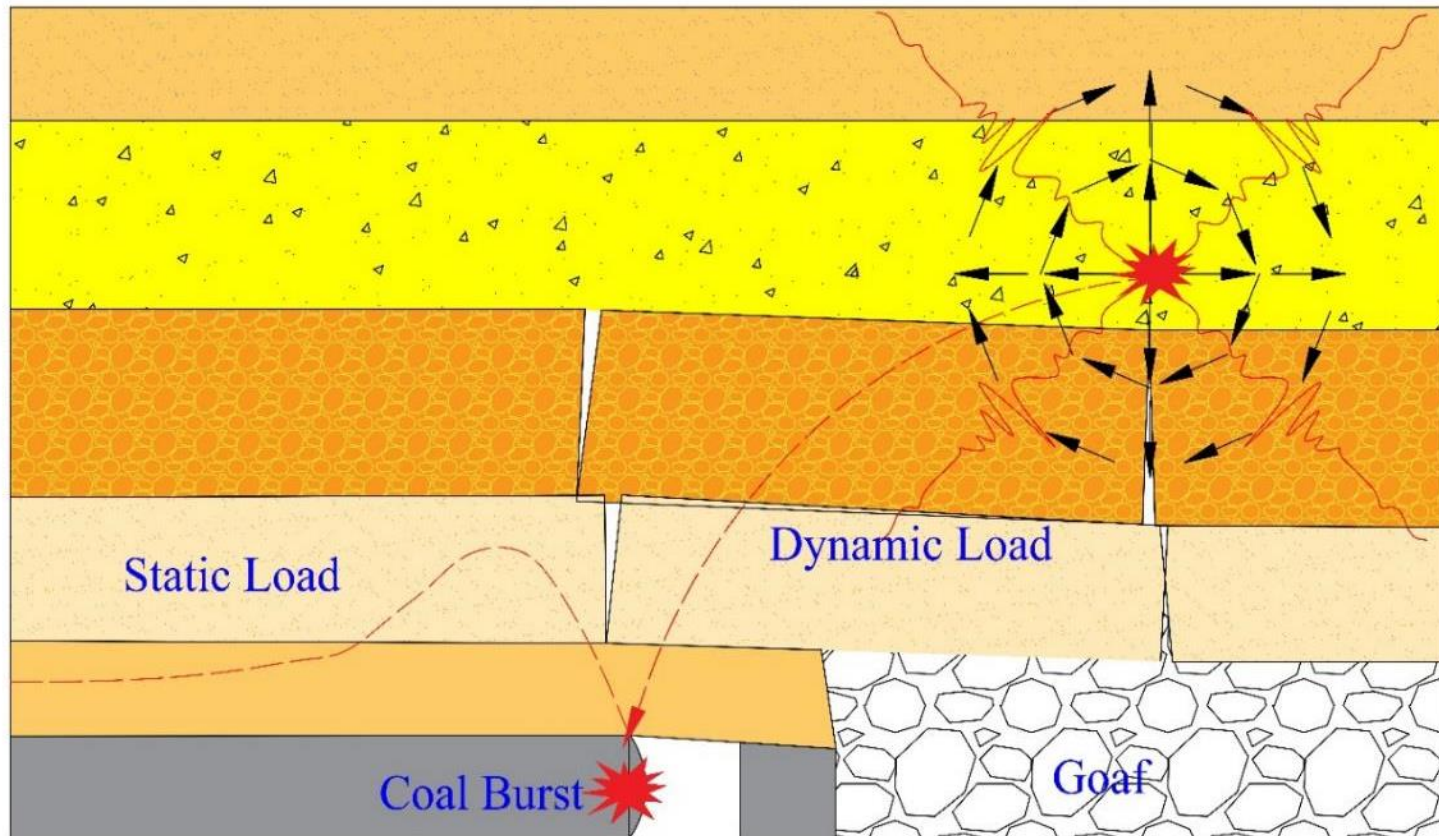


### Structural Geology of Coal Burst Sites

# Energy Analysis

## Static and Dynamic Load Superposition Theory

Coal burst will occur when the sum of static and dynamic load exceeds the minimum load required for coal burst formation. The energy released during coal burst is provided by static load and dynamic load.



Coal Burst Induced by Static and Dynamic Load superposition (Dou et al)

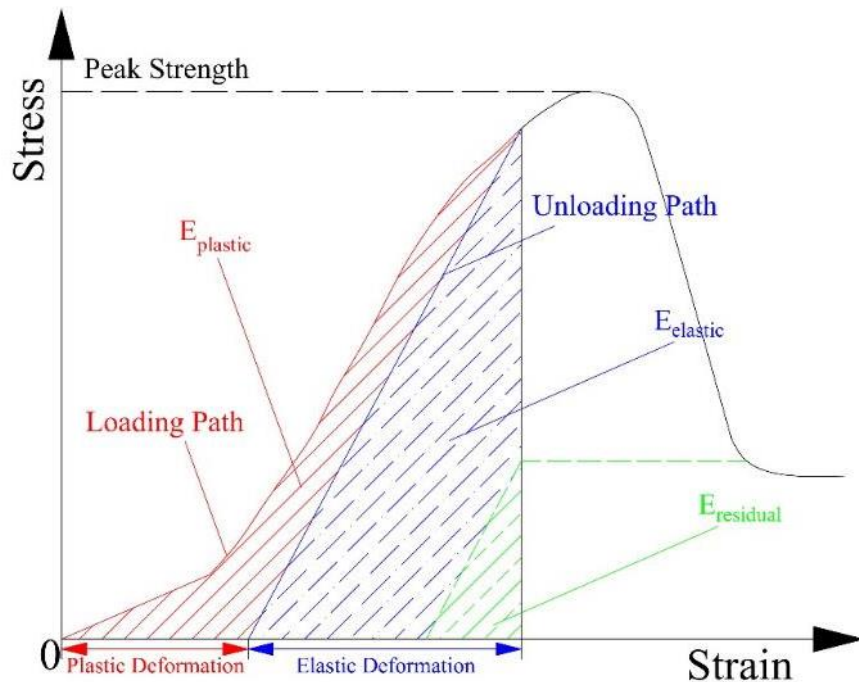


# Energy Analysis

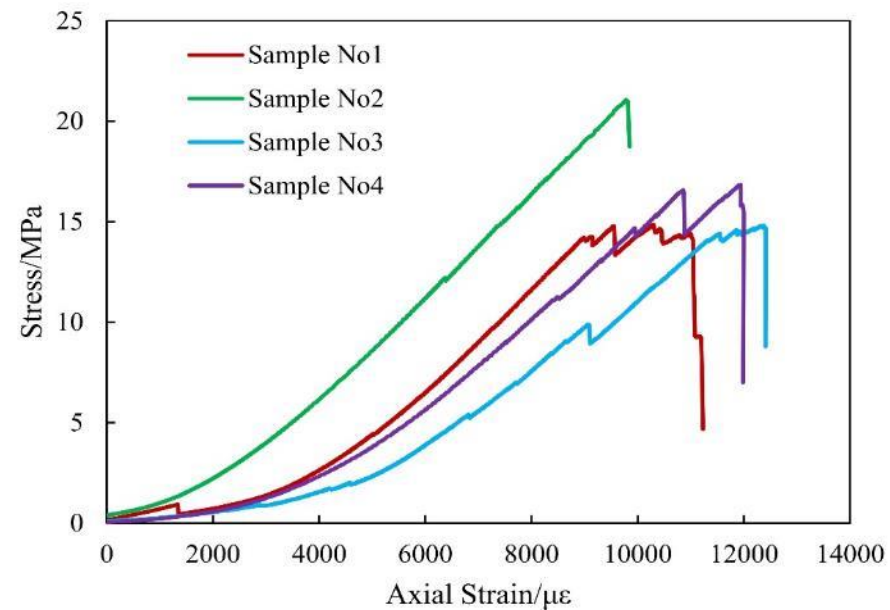
## Energy Dissipation Analysis

$$E_{plastic} + E_{elastic} = E_{total}$$

$$E_{elastic} = E_{crushing} + E_{kinetic} + E_{residual}$$



Schematic Diagram of Energy Accumulation before Peak Strength



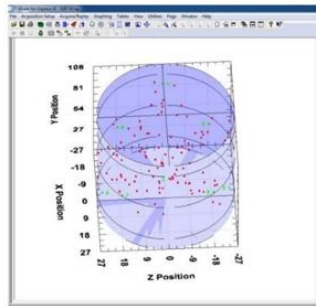
Stress versus Strain Curve of Coal Samples



# Energy Analysis

## Coal Burst Propensity Index

Coal burst propensity index method is an effective way to evaluate the burst risk of coal mines. Further tests with different coal seams are required in order to develop specific coal burst propensity classification method for Australian coal seams.



Locating of AE Sources

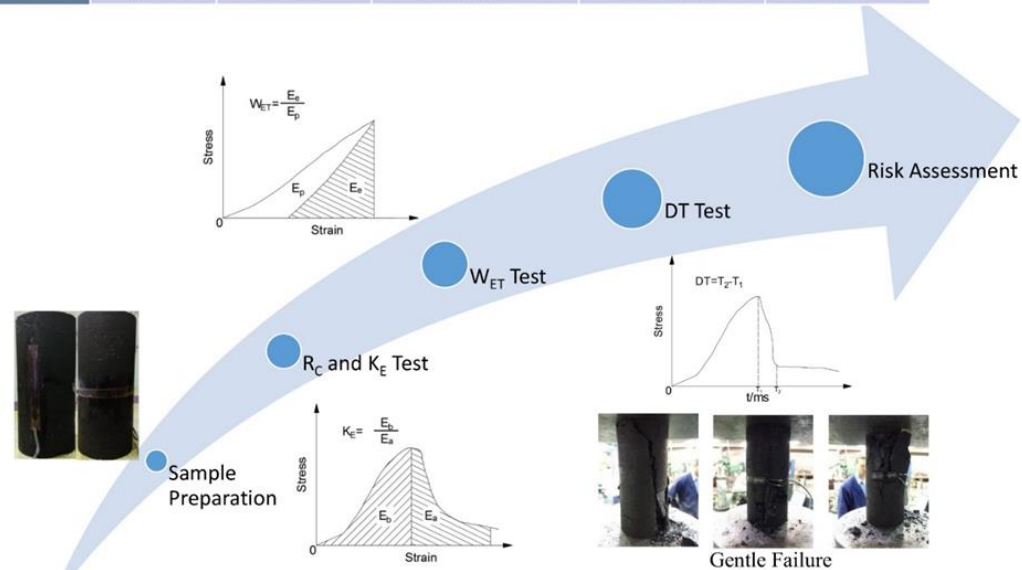


Loading Machine



Violent Failure

|                  |            | Risk Classification Form |                        |                       |                 |
|------------------|------------|--------------------------|------------------------|-----------------------|-----------------|
| Type             |            | I                        | II                     | III                   | IV              |
| Burst Propensity |            | None                     | Low                    | Moderate              | High            |
| Index            | DT/ms      | $DT > 10000$             | $1000 < DT \leq 10000$ | $500 < DT \leq 1000$  | $DT \leq 500$   |
|                  | $K_E$      | $K_E < 2$                | $2 \leq K_E < 3.5$     | $3.5 \leq K_E < 5$    | $K_E \geq 5$    |
|                  | $W_{ET}$   | $W_{ET} < 2$             | $2 \leq W_{ET} < 3.5$  | $3.5 \leq W_{ET} < 5$ | $W_{ET} \geq 5$ |
|                  | $R_C$ /Mpa | $R_C < 5$                | $5 \leq R_C < 10$      | $10 \leq R_C < 15$    | $R_C \geq 15$   |



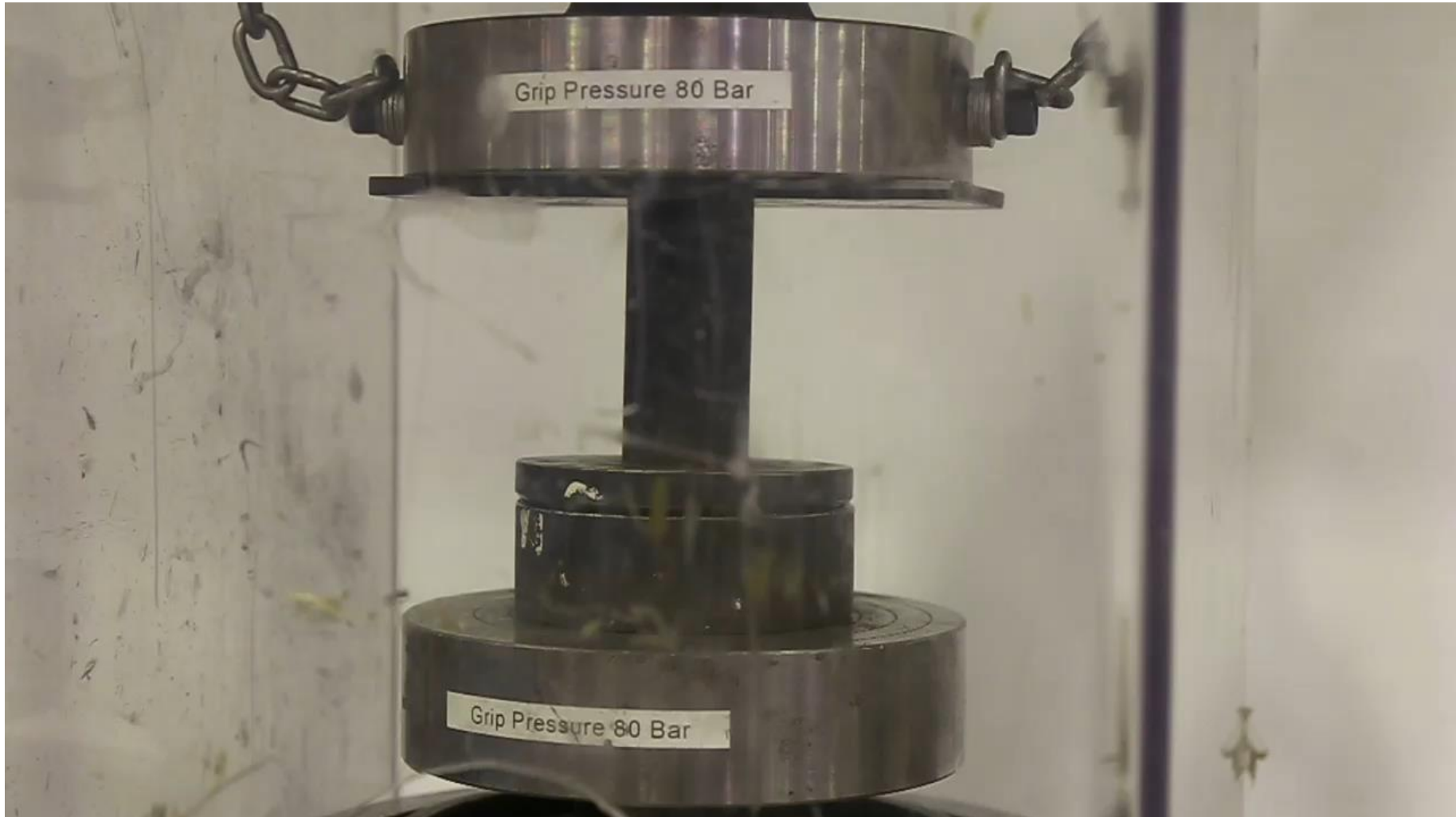
Coal Burst Propensity Index Test





# Energy Analysis

## Coal Burst Propensity Index





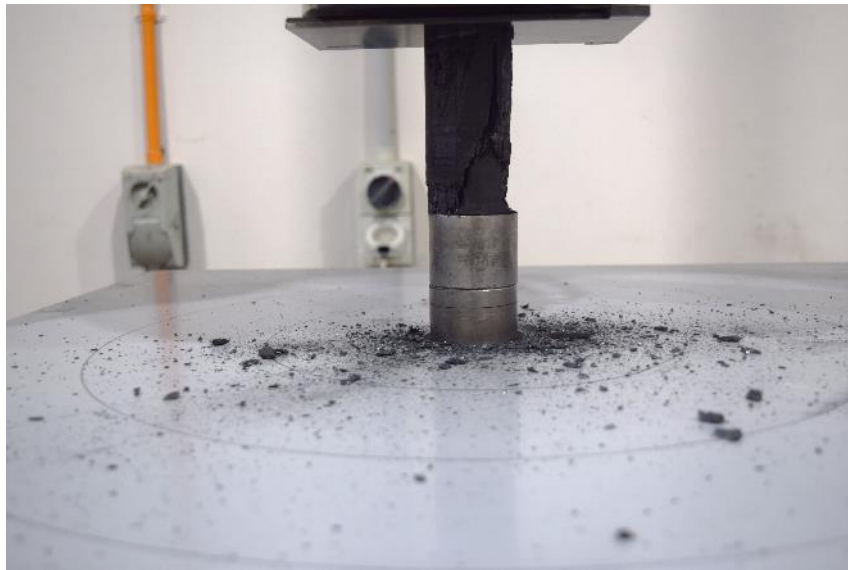
# Energy Analysis

## Kinetic Energy Estimation

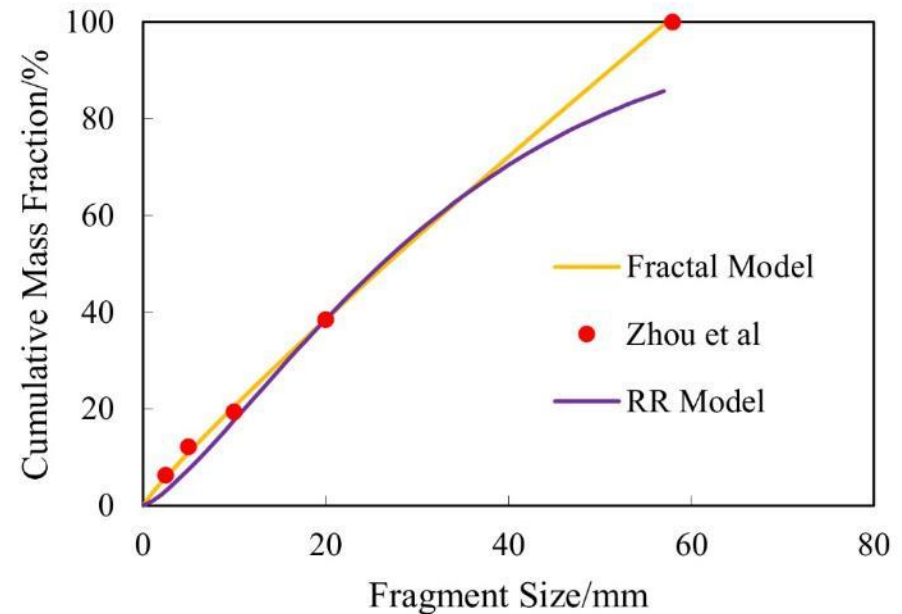
$$E_{elastic} = \frac{V}{2E_0} [\sigma_1^2 + \sigma_2^2 + \sigma_3^2 - 2\mu(\sigma_1\sigma_2 + \sigma_2\sigma_3 + \sigma_3\sigma_1)]$$

$$E_{kinetic} \cong E_{elastic} - E_{crushing}$$

$$F(d) = \left(\frac{d}{d_{max}}\right)^{(3-n)}$$



Coal Ejection Test



Fitting Functions of Fragment Size Distribution



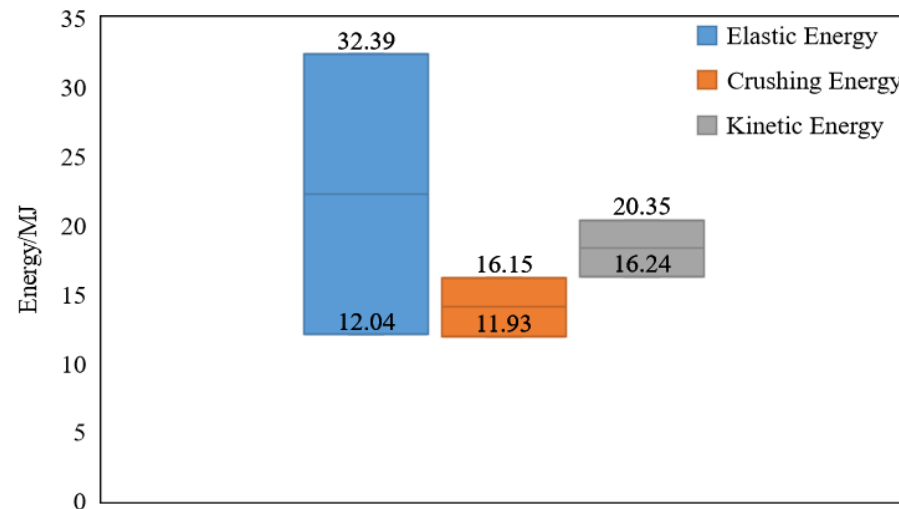
# Energy Analysis

## Kinetic Energy Estimation

The estimated kinetic energy carried by ejected coal is between 16.24 and 20.35 MJ. Considering the total mass of ejected coal, the average initial speed of ejected coal particles ranges from 24.98 to 27.96 m/s.

**Value of Main Parameters for Crushing Energy Estimation**

| Mining Depth | Stress Concentration Factor | Vertical Stress | Shape Factor | Density                | Volume of Ejected Coal | Weight of All Fragments | Rittinger Constant |
|--------------|-----------------------------|-----------------|--------------|------------------------|------------------------|-------------------------|--------------------|
| 555 m        | 1.75-2.87                   | 24.28-39.82 MPa | 1.5          | 1.37 g/cm <sup>2</sup> | 38 m <sup>3</sup>      | 52.06 t                 | 178.84 - 242.06    |

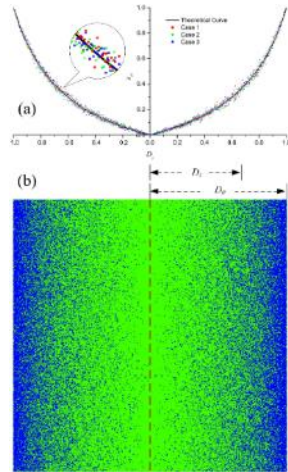


**Estimated Value of Kinetic Energy of Sidewall Burst**

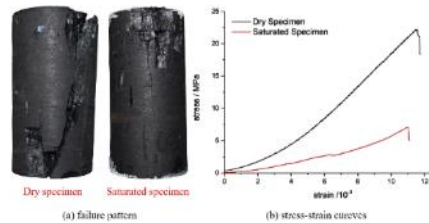
# Numerical Analysis

## -water effect on coal burst of pillar under geo-stress

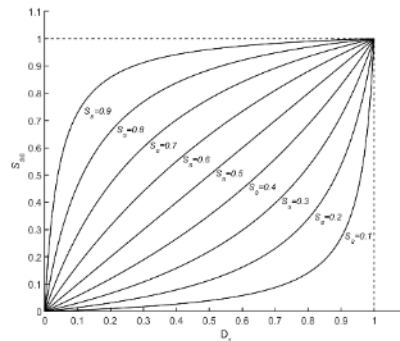
### Numerical model



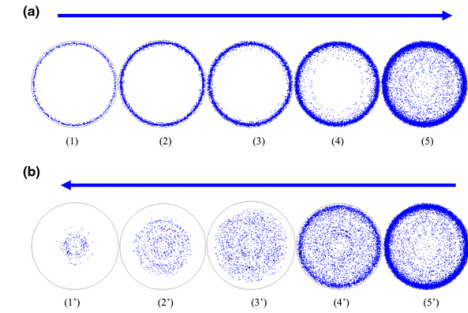
The water distribution curve and numerical model ( $s_c=0.3$ ); the blue patterns represent water-weakened contacts and the green patterns represent normal contacts.



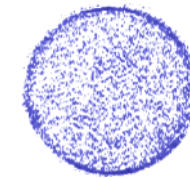
Comparison between experimental results of dry specimen and saturated specimen under uniaxial compression



The relationship between saturation degree and distance ratio: (a) saturation distribution; (b) evaporation distribution



Nuclear magnetic resonance (NMR)-images of sandstone disk with different water contents: a saturation process; b drying process (Zhou, 2016)



NMR-images of sandstone disk in saturation condition

Sectional saturation coefficient  $s_{ci}$

$$s_{ci} = m - \frac{m(1-m)}{D_r - m} \quad m < 0 \text{ or } m \geq 1$$

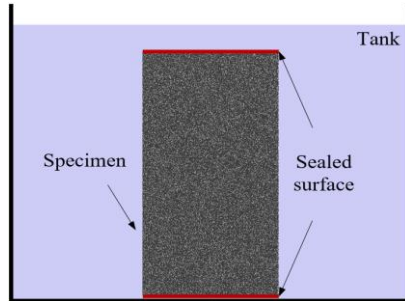
Overall water saturation coefficient  $s_c$

$$s_c = 2 \int_0^1 D_r \times s_{ci} dD_r = m + 2m(1-m) \left[ 1 + (1-m) \ln \left( \frac{m}{m-1} \right) \right]$$

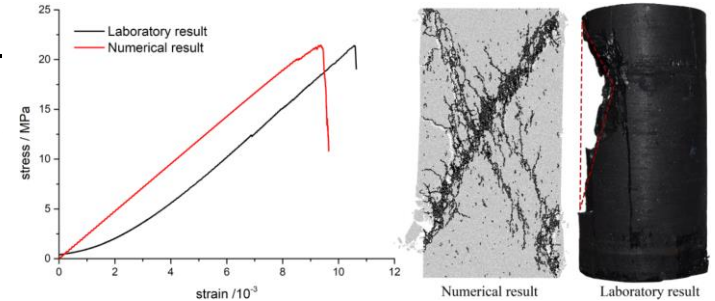


# Numerical Analysis

## Experiment preparation

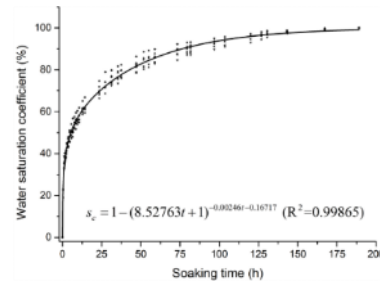
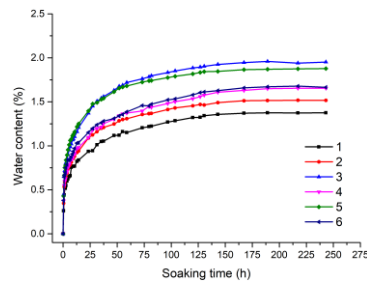


| Specimen No. | Group | Estimated saturation coefficient | Actual water content (%) | Length /mm | Diameter /mm |
|--------------|-------|----------------------------------|--------------------------|------------|--------------|
| D-1          | 1     | 0                                | 0.0                      | 108.24     | 54.02        |
| D-2          |       |                                  | 0.0                      | 108.03     | 53.83        |
| D-3          |       |                                  | 0.0                      | 107.94     | 53.74        |
| M-1          | 2     | 0.3                              | 0.47                     | 108.15     | 53.89        |
| M-2          |       |                                  | 0.62                     | 108.18     | 53.92        |
| M-3          |       |                                  | 0.58                     | 108.32     | 53.61        |
| H-1          | 3     | 0.7                              | 1.24                     | 107.87     | 53.87        |
| H-2          |       |                                  | 1.15                     | 108.26     | 53.93        |
| H-3          |       |                                  | 1.29                     | 108.07     | 53.75        |
| S-1          | 4     | 1.0                              | 1.66                     | 108.13     | 53.82        |
| S-2          |       |                                  | 1.67                     | 108.19     | 53.64        |
| S-3          |       |                                  | 1.88                     | 108.24     | 53.71        |



Schematic of soaking test for cylinder coal specimens and parameters of specimens for compression tests

Comparison between numerical and experimental results of dry specimen under uniaxial compression



Variation of water content and water saturation coefficient with time for coal specimens

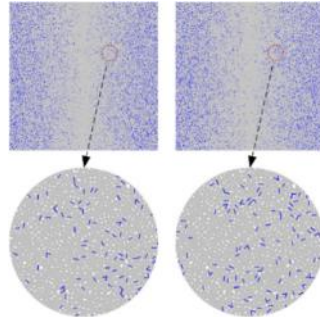
Mechanical properties of intact specimen in laboratory experiment and PFC numerical simulation

| Mechanical properties            | Experimental result | Numerical result | Deviation |
|----------------------------------|---------------------|------------------|-----------|
| Peak stress /MPa                 | 21.41               | 21.45            | 0.19%     |
| Young's modulus /GPa             | 2.43                | 2.39             | 1.65%     |
| Failure strain /10 <sup>-3</sup> | 10.57               | 9.36             | 11.45%    |

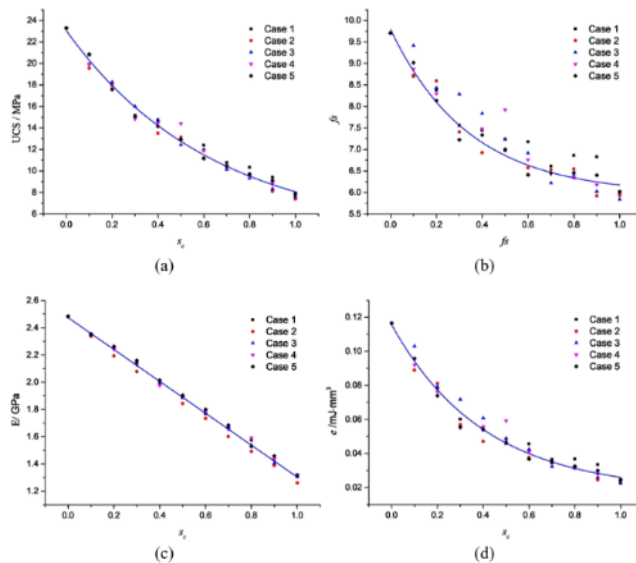


# Numerical Analysis

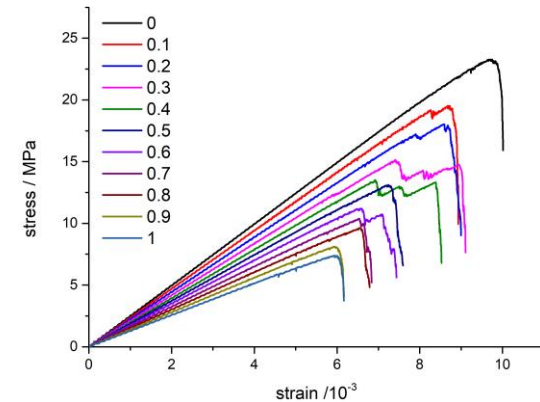
## Parameter calibration



Comparison between two numerical models with the same saturation coefficient



Relationships between water saturation coefficient and mechanical properties in numerical simulations: (a) UCS; (b) Failure strain; (c) Young's modulus; (e) Strain energy per unit volume



Stress-strain curves for specimens with different saturation coefficients in case 2

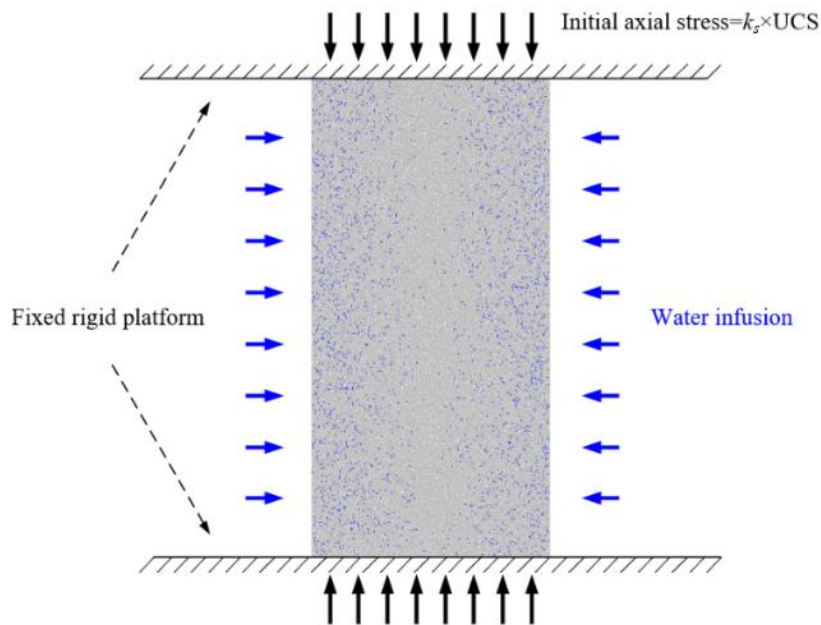
The fitting functions between mechanical properties and water saturation coefficient for simulation results

| Mechanical properties               | Fitting function                         | R <sup>2</sup> |
|-------------------------------------|--|----------------|
| UCS $\sigma_c$                      | $\sigma_c = 4.142 + 18.910e^{-1.573s_c}$ | 0.9961         |
| Failure strain $f_s$                | $f_s = 5.965 + 3.825e^{-2.905s_c}$       | 0.9770         |
| Young's modulus $E$                 | $E = 2.475 - 1.171s_c$                   | 0.9982         |
| Absorbed energy per unit volume $e$ | $e = 0.037 + 0.067e^{-2.650s_c}$         | 0.9997         |

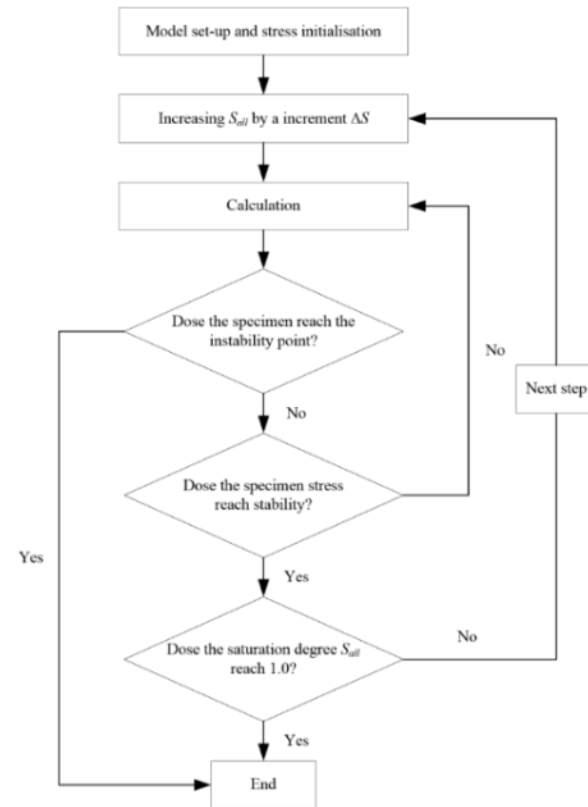


# Numerical Analysis

## Numerical simulation



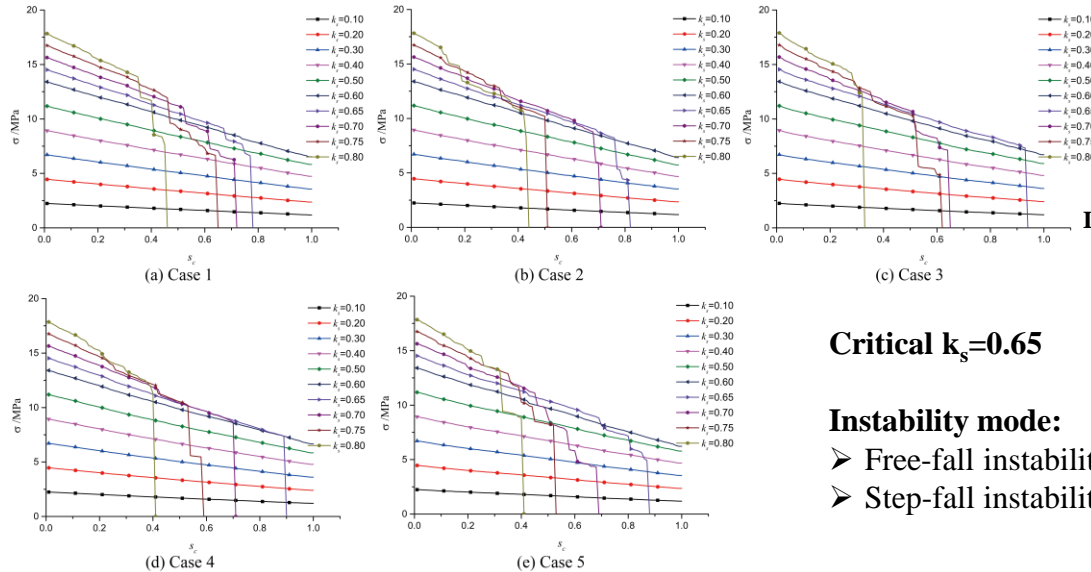
Sketch of the numerical experiment



Flow chart for the simulation procedure

# Numerical Analysis

## Stress evolution

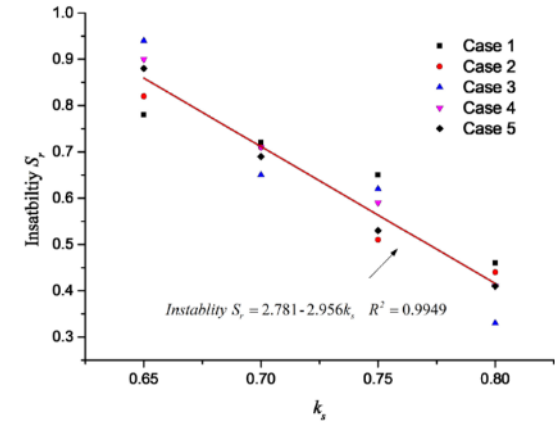


Stress evolution with the increase of water saturation coefficient  $s_w$  under different initial stress conditions

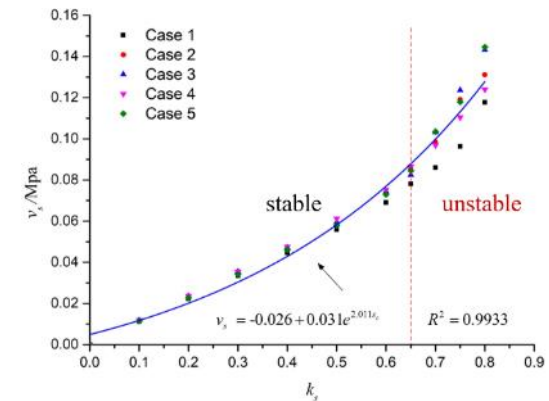
Critical  $k_s=0.65$

Instability mode:

- Free-fall instability
- Step-fall instability



Instability water saturation coefficient for specimens in high-stress conditions



$v_s$  evolution curves with  $k_s$  increasing

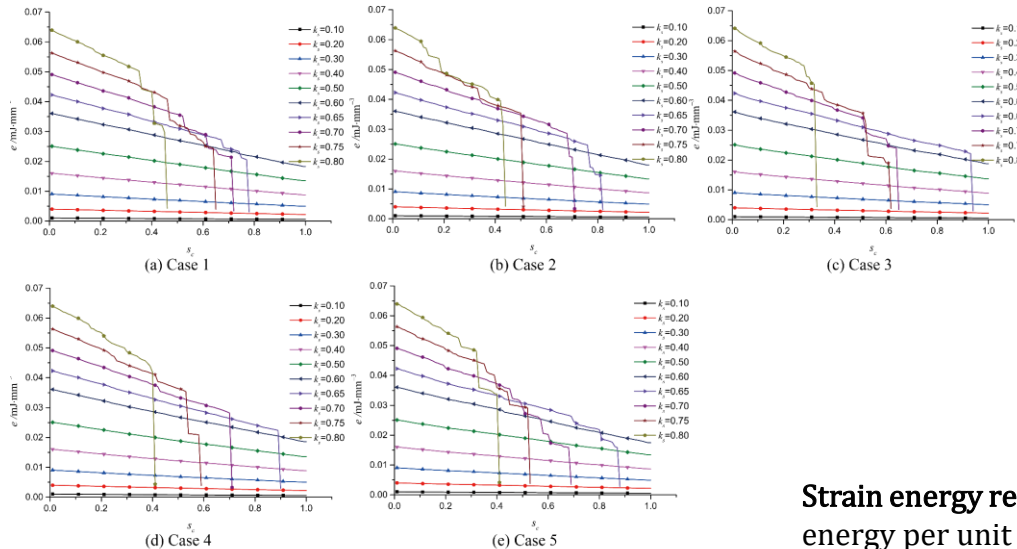
**Stress energy releasing rate  $v_s$ :** the decrement of axial stress when the water saturation coefficient increased 1%





# Numerical Analysis

## Energy evolution

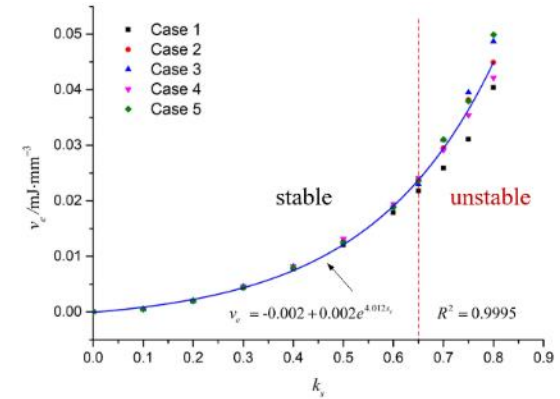


Strain energy evolution with the increase of water saturation coefficient under different initial stress conditions

Strain energy per unit volume  $e$

$$e = \frac{W}{V}$$

$W$  is the total work done by the testing system before the instability point of a specimen,  $V$  is the volume of the specimen



ve evolution curves with ks increasing

Strain energy releasing rate  $v_e$ : the decrement of released strain energy per unit volume when the water saturation increased 1%

Initial stress coefficient:

**65%~80% UCS:** Lower instability point and higher coal burst risk.

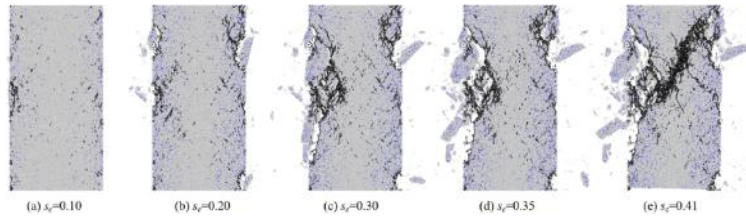
**40%~65% UCS:** Water infusion is an effective approach to reduce coal burst risk as having been reported by many literatures.

**≤40% UCS:** Water has limited effect on releasing stress and energy for coal.



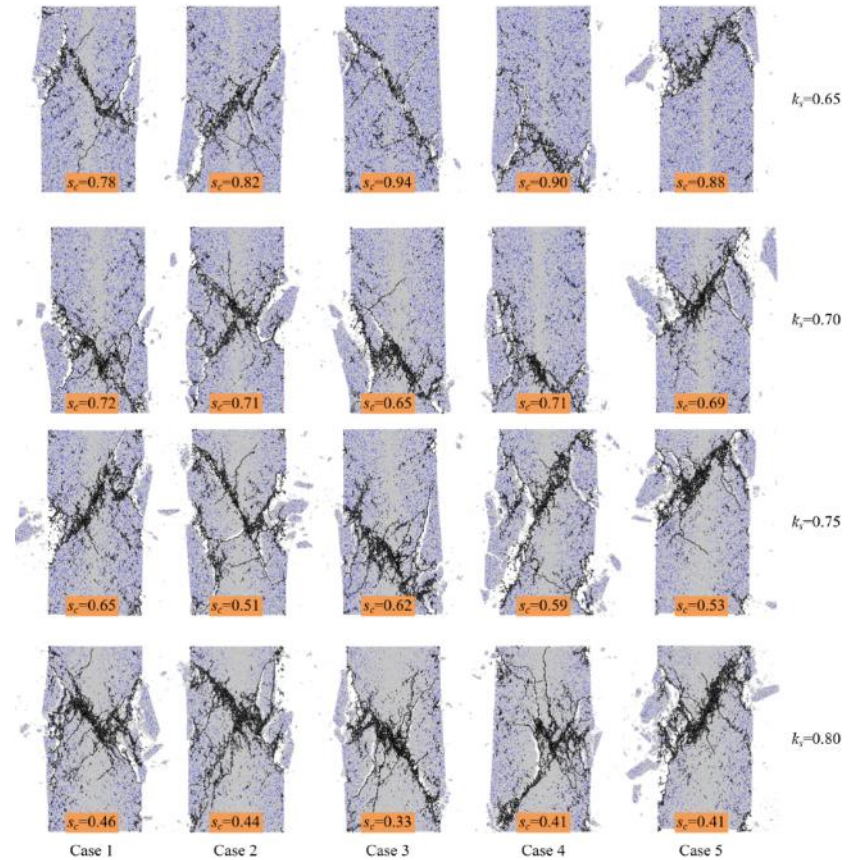
# Numerical Analysis

## Failure mode



Failure evolution of specimen in Case 5,  $k_s=0.8$

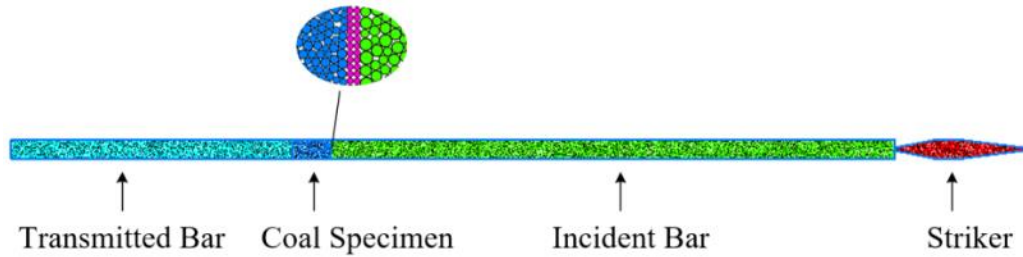
- Similar failure patterns
- Splitting failure in water-rich area
- Shear-dominated failure



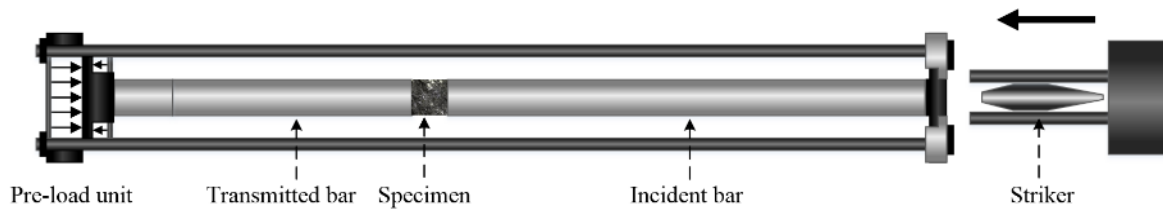
Final failure patterns of all damaged specimens

# Current Work

## Numerical Modelling of Dynamic Load



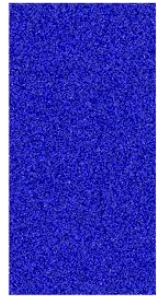
Numerical model of SHPB test system



Numerical model of Pre-load SHPB test system



Drop hammer test system





# Current Work

## Protective Structure on CM





# Conclusions

## Energy Analysis

1. The main energy source of coal burst is provided by static load.
2. Coal burst propensity index can evaluate the coal burst risk by reflecting the energy accumulation and dissipation behavior.
3. The average ejection velocity of coal particles from roadway sidewall can reach 24.98-27.96 m/s.

## Numerical modeling of pillar instability

### 1. Instability

Free-fall instability : stress and energy decreased linearly and stably and then overall instability appeared suddenly.

Step-fall instability : several times of stress and energy drop and had been damaged obviously before the final instability.

2. The axial stress and strain energy within the specimens are more sensitive to water under a higher initial axial stress condition.
3. The stress releasing rate  $v_s$  and energy releasing rate  $v_e$  are suggested to be an effective index to assess the stability and of pillar.



Questions?