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Solution of Two-dimensional Linear and Nonlinear Unsteady Schrödinger Equation using "Quantum Hydrodynamics" Formulation with a MLPG Collocation Method

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Abstract: A numerical solution of the linear and nonlinear time-dependent Schrö-7 dinger equation is obtained, using the strong form MLPG Collocation method. 8 Schrödinger equation is replaced by a system of coupled partial differential equaq tions in terms of particle density and velocity potential, by separating the real and 10 imaginary parts of a general solution, called a quantum hydrodynamic (QHD) equa-11 tion, which is formally analogous to the equations of irrotational motion in a clas-12 sical fluid. The approximation of the field variables is obtained with the Moving 13 Least Squares (MLS) approximation and the implicit Crank-Nicolson scheme is 14 used for time discretization. For the two-dimensional nonlinear Schrödinger equa-15 tion, the lagging of coefficients method has been utilized to eliminate the non-16 linearity of the corresponding examined problem. A Type-I nodal distribution is 17 used in order to provide convergence for the discrete Laplacian operator used at the 18 governing equation. Numerical results are validated, comparing them with analyti-19 cal and numerical solutions. 20

Keywords: MLPG Collocation Method, Moving Least Squares, Schrödinger Equa tion, Quantum Hydrodynamics.

23 1 Introduction

The meshless (or meshfree) methods are being actively developed as a powerful numerical tool for various engineering and physical applications. The primary reason for the significant interest in meshless computational procedures is that most of the established numerical techniques, such as the Finite Element Method (FEM), the Finite Volume Method (FVM), the Finite Difference Method (FDM) and the

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Spectral Method (SP) require a mesh. The automatic generation of a good quality 29 mesh poses a significant problem in the analysis of practical engineering systems. 30 Moreover, the simulation and the analysis of certain types of problems (like dy-31 namic crack propagation, pulsatile and transient flows) often require an expensive 32 remeshing operation. Meshless techniques overcome these difficulties, associated 33 with the meshing and re-meshing procedures, by eliminating the mesh altogether. 34 Interpolation is performed in terms of nodal points scattered at the spatial domain 35 using functions having compact support. A weighted residual technique is used to 36 generate the discrete set of equations corresponding to the governing partial differ-37 ential equations [Liu (2003), Liu and Gu (2005)]. 38 Since the meshless methods emerged as a potential alternative for solutions in com-39

putational mechanics, a variety of such approaches have appeared. Over the last 40 decades, several meshfree methods have been proposed since the prototype of the 41 meshfree methods, the Smoothed Particle Hydrodynamics (SPH), was born [Gin-42 gold and Monaghan (1977)]. These methods include the Diffuse Approximation 43 Method (DAM) [Nayroles, Touzot and Villon (1991)], that is closely related to 44 the Moving Least Squares method; the Diffuse Element Method (DEM) [Nayroles, 45 Touzot and Villon (1992)], developed by the Moving Least Squares approximation, 46 and the Element Free Galerkin method (EFG) [Lu, Belytschko and Gu (1994)]; 47 the Reproducing Kernel Particle Method (RKPM) [Liu, Jun and Zhang (1995), 48 Liu, Jun, Li, Adee and Belytschko (1995)], which is used to improve the SPH 49 approximation; the Partition of Unity Finite Element Method (PUFEM) [Melenk 50 and Babuska (1996)]; the hp-Clouds [Duarte and Oden (1996)]; the Moving Least-51 Square Reproducing Kernel Method (MLSRK) [Liu, Li and Belytschko (1996)]; 52 the meshless Local Boundary Integral Equation Method (LBIE) [Zhu, Zhang and 53 Atluri (1998)]; the Meshless Local Petrov-Galerkin method (MLPG) [Atluri, Kim 54 and Cho (1999), Atluri and Shen (2002)]; the Finite Point method (FPM) [Onate, 55 Idelsohn, Zienkiewicz and Taylor (1995)]; the meshless point collocation methods 56 (MPC) [Aluru (2000)], and more. 57

The present paper is referred to the numerical computation of the two-dimensional (2D) time-dependent Schrödinger equation. Linear Schrödinger equation is written as

$$-i\frac{\partial\psi}{\partial t} = \frac{\partial^{2}\psi}{\partial x^{2}} + \frac{\partial^{2}\psi}{\partial y^{2}} + V(x,y)\psi, \quad (x,y) \in \Omega, \quad 0 \le t \le T$$
(1a)

in some continuous domain with suitable initial Dirichlet and Neumann boundary conditions and an arbitrary potential function V(x,y). The corresponding initial condition is given by

$$\Psi(x, y, 0) = h(x, y, 0), \quad (x, y) \in \Omega$$
(1b)

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and the boundary conditions by

$$\Psi(x, y, t) = s(x, y, t), \quad (x, y) \in \partial \Omega^D, \quad 0 \le t \le T$$
(1c)

$$\frac{\partial \psi}{\partial \boldsymbol{n}}(x, y, t) = g(x, y, t), \quad (x, y) \in \partial \Omega^N, \quad 0 \le t \le T$$
(1d)

where $i = \sqrt{-1}$ is the unit imaginary number, *T* is the final time, *h*, *s* and *g* are known functions, and $\partial \Omega = \partial \Omega^D \cup \partial \Omega^N$, where $\partial \Omega^D$ and $\partial \Omega^N$ are the Dirichlet

and the Neumann parts of the boundary $\partial \Omega$ and *n* is the unit outward vector to $\partial \Omega$.

This type of partial equation models many physical problems and find applications in quantum mechanics and various quantum dynamics calculations [Arnold (1998), Hajj (1985), Ixaru (1997)], in electromagnetic wave propagation and the design of certain optoelectronic devices [Levy, (2000), Huang, Xu, Chu and Chaudhuri (1992)], and finally, in underwater acoustics [Tappert (1977)]. The time-dependent Schrödinger equation can be represented in a hydrodynamical form, called a quantum hydrodynamic (QHD) equation, a formulation which is analogous to the equations of irrotational motion in a classical fluid [Gasser, Lin and Markowich (2000), Kalita, Chhabra and Kumar (2006)]. In this formulation, system (1) is replaced by a system of partial differential equations in terms of particle density and velocity potential, by separating the real and imaginary parts of a general solution

$$-\frac{\partial u}{\partial t} = \nabla^2 \upsilon + V \upsilon,$$

$$\frac{\partial \upsilon}{\partial t} = \nabla^2 u + V u,$$

(2)

obtained by expressing ψ as $\psi = u + iv$, where u and v are real-values functions.

There have been numerous attempts to develop numerical schemes for equations (1) 62 or the system (2). In [Simos (2008), Simos (2007)] trigonometrically-fitted meth-63 ods were utilized for the numerical solution of the Schrödinger equation. The au-64 thors of [Kalita, Chhabra and Kumar (2006), Subasi (2002)] studied models similar 65 to the present problem using finite-difference techniques. Finite-difference meth-66 ods are well-known as the first technique for solving partial differential equations 67 (PDEs). In [Dehghan (2002)] explicit finite difference methods were used for solv-68 ing the governing equations, while in [Dehghan (1999)] the need of using a large 69 amount of CPU time in implicit finite-difference schemes limit the applicability of 70 these methods. Furthermore, these methods provide the solution of the problem 71 on mesh points only, and the accuracy of the techniques is reduced in non-smooth 72 and non-regular domains. Thus, alternative computational methods, such as global 73 Radial Basis Functions [Dehghan and Shokri (2007)], were used for the numerical 74 solution of the Eq. (1). 75

In the present paper we investigate a different approach to find the solution of linear 76 and nonlinear Schrödinger equation. We present a numerical scheme to solve the 77 two-dimensional (2D) time-dependent Schrödinger equation using the Collocation 78 method, while we approximate the solution directly using Moving Least Squares. 79 Actually, the meshless point collocation (MPC) method is a case of MLPG when 80 the collocation Dirac's Delta function is used as the test function [Atluri and Shen 81 (2002)]. To test the robustness, the accuracy and the efficiency of the proposed 82 scheme, it is applied to four examples having analytical solutions, with our results 83 exhibiting very good agreement with the analytical ones. Additionally, our results 84 are compared with a meshless collocation and radial basis function method using 85 multiquadrics (MO) and the Thin Plate Splines (TPS). The layout of the paper is 86 as follows. In Section 2 we present the methodology for the implementation of 87 the Moving Least Squares approximation for the solution of OHD equations. In 88 Section 3 we apply this technique on the two-dimensional (2D) time-dependent 80 Schrödinger equation. The results of the numerical experiments are presented in 90 Section 4, while Section 5 is dedicated to a brief conclusion. 91

92 2 Moving Least Squares Approximation

93 2.1 Methodology

In the moving least-squares technique, the approximation $u^{h}(\mathbf{x})$ is expressed as the inner product of a vector of the polynomial basis, $\mathbf{p}(\mathbf{x})$ and a vector of the coefficients, $\mathbf{a}(\mathbf{x})$

$$u^{h}(\mathbf{x}) = \mathbf{p}^{T}(\mathbf{x}) \, \mathbf{a}(\mathbf{x}), \tag{3}$$

where $p(x) \in \mathbb{R}^m$, $a(x) \in \mathbb{R}^m$ and *m* is the number of monomials in the polynomial basis (in the present study *m*=6). The local character of the moving least-squares (MLS) approximation can be viewed as a generalization of the traditional leastsquares approximation, in which the vector **a** is not a function of **x**.

Equation (3) is commonly referred to as the global least-squares approximation. In addition, there exists a unique local approximation associated with each point in the domain. In order to determine the form of a(x), a weighted discrete error norm,

$$J(\mathbf{x}) = \sum_{I=1}^{n} w_I(\mathbf{x}) \left[\sum_{j=1}^{m} p_j^T(\mathbf{x}_I) \, \boldsymbol{a}(\mathbf{x}) - u_i \right]^2 \tag{4}$$

is constructed and sequentially minimized. Here, $w_I(\mathbf{x})$ denotes the weight function, $w_I(\mathbf{x}) \equiv w(\mathbf{x} - \mathbf{x}_I)$, associated with node *I*, and the quantity in brackets is the difference between the local approximation at node *I* and the data at nodes *I*, that

¹⁰¹ is u_i , and n is the number of nodes in the support of $w_I(\mathbf{x})$. The minimization of ¹⁰² Eq.(4) with respect to $\mathbf{a}(\mathbf{x})$ determines $\mathbf{a}(\mathbf{x})$. The local approximation associated ¹⁰³ with point \mathbf{x} is used only in the minimization process and is equivalent to the global ¹⁰⁴ approximation at the single point \mathbf{x} . Compact support of the weight functions gives ¹⁰⁵ the moving least-squares method its local character.

106 2.2 Shape functions and their derivatives

The minimization of Eq. (4),

$$\frac{\partial J(\mathbf{x})}{\partial \mathbf{a}(\mathbf{x})} = 0$$

results in the linear system

$$\boldsymbol{A}(\boldsymbol{x})\boldsymbol{a}(\boldsymbol{x}) = \boldsymbol{B}(\boldsymbol{x})\boldsymbol{U}_{\boldsymbol{s}},$$

where U_s is a vector containing the nodal data, $U_s^T = [u_1, u_2, ..., u_n]$, and

$$\mathbf{A}(\mathbf{x}) = \sum_{I=1}^{n} w_I(\mathbf{x}) \mathbf{p}(\mathbf{x}_I) \mathbf{p}^T(\mathbf{x}_I),$$
(7)

$$\boldsymbol{B}(\boldsymbol{x}) = \begin{bmatrix} w_1(\boldsymbol{x}) \boldsymbol{p}(\boldsymbol{x}_1) & w_2(\boldsymbol{x}) \boldsymbol{p}(\boldsymbol{x}_2) & \dots & w_n(\boldsymbol{x}) \boldsymbol{p}(\boldsymbol{x}_n) \end{bmatrix},$$
(8)

where $A \in \mathbb{R}^{m \times m}$ and $B \in \mathbb{R}^{m \times n}$. The matrix A must be inverted at every sampling point. Substitution of the solution of (Eq.(6)) into the global approximation (Eq.(3)), completes the least-squares approximation,

$$u^{h}(\mathbf{x}) = \underbrace{p^{-1}(\mathbf{x}) A^{-1}(\mathbf{x}) B^{\prime}(\mathbf{x})}_{\phi(x)} U_{s}^{\prime}.$$
(9)

Here, the spatial dependence has been lumped into one row matrix, $\varphi(\mathbf{x})$ and, therefore, the approximation takes the form of a product of a matrix of shape functions with a vector of nodal data. Derivatives of the shape functions may be calculated by applying the product rule to

$$\boldsymbol{\varphi} = \boldsymbol{p}^T \boldsymbol{A}^{-1} \boldsymbol{B}. \tag{10}$$

In order to obtain the spatial derivatives of the approximation function, $u^{h}(\mathbf{x})$, it is necessary to obtain the derivatives of the MLS shape functions, $\varphi_{i}(\mathbf{x})$,

$$\frac{\partial}{\partial x_j} u^h(\mathbf{x}) = \frac{\partial}{\partial x_j} \sum_{i=1}^n \varphi_i(\mathbf{x}) u_i = \sum_{i=1}^n \left\{ \frac{\partial}{\partial x_j} \varphi_i(\mathbf{x}) \right\} u_i, \quad x_j = x, y, z.$$
(11)

5

(5)

(6)

The derivative of the shape function is given as

$$\frac{\partial \phi(\mathbf{x})}{\partial x_j} = \frac{\partial \left(\mathbf{p}^T \mathbf{A}^{-1} \mathbf{B}_i \right)}{\partial x_j} = \frac{\partial \mathbf{p}^T}{\partial x_j} \mathbf{A}^{-1} \mathbf{B}_i + \mathbf{p}^T \frac{\partial \left(\mathbf{A}^{-1} \right)}{\partial x_j} \mathbf{B}_i + \mathbf{p}^T \mathbf{A}^{-1} \frac{\partial \mathbf{B}_i}{\partial x_j}, \quad x_j = x, y, z$$
(12)

where $\frac{\partial(\mathbf{A}^{-1})}{\partial x_j} = -\mathbf{A}^{-1}(\mathbf{x})\mathbf{A}_{,j}(\mathbf{x})\mathbf{A}^{-1}(\mathbf{x})$. Regarding the second order derivative of the unknown function we get

$$\frac{\partial^{2} \varphi(x)}{\partial x_{j}^{2}} = \frac{\partial}{\partial x_{j}} \left(\frac{\partial \varphi(x)}{\partial x_{j}} \right) = \frac{\partial}{\partial x_{j}} \left(\frac{\partial \mathbf{p}^{T}}{\partial x_{j}} \mathbf{A}^{-1} \mathbf{B}_{i} + \mathbf{p}^{T} \frac{\partial (\mathbf{A}^{-1})}{\partial x_{j}} \mathbf{B}_{i} + \mathbf{p}^{T} \mathbf{A}^{-1} \frac{\partial \mathbf{B}_{i}}{\partial x_{j}} \right)$$

$$= \frac{\partial^{2} \mathbf{p}^{T}}{\partial x_{j}^{2}} \mathbf{A}^{-1} \mathbf{B}_{i} + \frac{\partial \mathbf{p}^{T}}{\partial x_{j}} \frac{\partial (\mathbf{A}^{-1})}{\partial x_{j}} \mathbf{B}_{i} + \frac{\partial \mathbf{p}^{T}}{\partial x_{j}} \mathbf{A}^{-1} \frac{\partial \mathbf{B}_{i}}{\partial x_{j}} + \frac{\partial \mathbf{p}^{T}}{\partial x_{j}} \frac{\partial (\mathbf{A}^{-1})}{\partial x_{j}} \mathbf{B}_{i} + \mathbf{p}^{T} \frac{\partial^{2} (\mathbf{A}^{-1})}{\partial x_{j}^{2}} \mathbf{B}_{i} + \mathbf{p}^{T} \frac{\partial (\mathbf{A}^{-1})}{\partial x_{j}} \frac{\partial \mathbf{B}_{i}}{\partial x_{j}} + \frac{\partial \mathbf{p}^{T}}{\partial x_{j}} \frac{\partial (\mathbf{A}^{-1})}{\partial x_{j}} \frac{\partial \mathbf{B}_{i}}{\partial x_{j}} + \mathbf{p}^{T} \frac{\partial (\mathbf{A}^{-1})}{\partial x_{j}} \frac{\partial \mathbf{B}_{i}}{\partial x_{j}} + \mathbf{p}^{T} \mathbf{A}^{-1} \frac{\partial^{2} \mathbf{B}_{i}}{\partial x_{j}^{2}},$$
(13)

where
$$x_j = x, y, z$$
 and $\frac{\partial^2(\mathbf{A}^{-1})}{\partial x_j^2} = -\frac{\partial(\mathbf{A}^{-1})}{\partial x_j}\mathbf{A}\mathbf{A}^{-1} - \mathbf{A}^{-1}\frac{\partial\mathbf{A}}{\partial x_j}\mathbf{A}^{-1} - \mathbf{A}^{-1}\mathbf{A}\frac{\partial(\mathbf{A}^{-1})}{\partial x_j}.$

Weight Function 2.3 108

The weight function is non-zero over a small neighborhood of x_i , called the support domain of node *i*. The choice of the weight function $W(\mathbf{x} - \mathbf{x}_i)$ affects the resulting approximation $u^h(\mathbf{x}_i)$ inherently. In the present paper a Gaussian weight function is used [Liu (2003), Bourantas, Skouras and Nikiforidis (2009)], yet the support domain does not have a standard point density value. Instead, a constant number of nodes are used for the approximation of the field function.

$$W(\boldsymbol{x} - \boldsymbol{x}_i) \equiv W(d) = \left\{ \begin{array}{c} e^{-\left(\frac{d_I}{a}\right)^2} \\ 0 \end{array} \right\},\tag{14}$$

where I = 1, 2, 3, ..., q are the nodes that produce the support domain of node x_i , 109 and $d = \frac{|\mathbf{x} - \mathbf{x}_i|}{a_0^2}$ with a_0 a prescribed constant (often $a_0 = 0.2$). 110

111 3 Collocation formulation

112 3.1 General description

The Meshless Point Collocation method is a MFree "strong-form" description me-113 thod. The "strong-form" of the governing equations and the boundary conditions 114 are used and discretized by collocation techniques. The aforementioned formula-115 tions possess the following attractive advantages. They are truly meshless and the 116 implementing procedure is straightforward, while the algorithms and the imple-117 mentation can be kept simple, particularly when handling problems with Dirich-118 let boundary conditions solely. Under these conditions, these methods are highly 119 efficient computationally, even with the application of polynomial approximation 120 functions, and the solution can be systematically obtained with increased accu-121 racy, compared to FEM, FVM, FDM, or other computational methods. In general, 122 MFree strong-form methods may still suffer from some local stability and accu-123 racy issues, depending on the problem [Liu and Gu (2005)]. However, these local 124 restrictions are now systematically avoided with the utilization of specific nodal 125 distributions (Type-I) and proper local point cloud refinement procedures, in ac-126 cordance with [Bourantas, Skouras and Nikiforidis (2009), Kim and Liu (2006)], 127 even for natural or mixed type boundary conditions. The robustness of these meth-128 ods has, however, been an issue especially for scattered set of points. The stability 129 and the convergence of the collocation methods are ensured by the resulting lin-130 ear or linearized algebraic system. If the latter possesses some attractive features 131 then both the stability and the convergence are ensured. In fact, the robustness of 132 the collocation methods can be improved by understanding the possible sources of 133 errors. Specifically, the errors could arise because of the way the meshless approx-134 imation functions and their derivatives have been constructed for a scattered point 135 of points or because of the way the discretization of the governing equations has 136 been performed. When the meshless approximation functions and its derivatives do 137 not satisfy certain conditions (referred to as the positivity conditions) for a given 138 point distribution, it is possible to get large numerical errors when using colloca-139 tion methods. To satisfy the positivity conditions, the weighting function used in 140 the construction of the approximation functions can play an important role. These 141 studies suggest that positivity conditions can be important when using meshless 142 collocation methods. Additionally, the convergence of the discrete Laplacian oper-143 ator for Dirichlet boundary conditions has been proved when a regular grid (named 144 Type-I) is used. Thus, both the stability and the convergence of the meshless point 145 collocation method, using MLS approximation and regular nodal distribution are 146 ensured. 147

¹⁴⁸ Collocation method using MLS may be considered as a special case of the "weak-

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form" methods [Atluri, (2004)]. Moreover, this collocation method may be consid-149 ered as a "weak-solution", with a Dirac delta function as the test (weight) function 150 [Atluri, Liu and Han (2006)]. The weighted residual method provides a flexible 151 mathematical framework for the construction of a variety of numerical solution 152 schemes for the differential equations arising in the field of both science and en-153 gineering. Its application, in conjunction with the Moving Least Square (MLS) 154 approximation method, yields powerful solution algorithms for the governing equa-155 tions. 156

157 3.2 Time-dependent Meshless Point Collocation method

The collocation scheme using the Moving Least Squares approximation used in the present work and applied for the spatial discretization of the unsteady homogeneous diffusion equation will be discussed next, along with the explicit Euler, θ -weighted time-stepping scheme used for temporal discretization.

Consider the governing equations of the unsteady problem

$$-\frac{\partial u}{\partial t} = \nabla^2 \upsilon + V \upsilon, \tag{15}$$

$$\frac{\partial v}{\partial t} = \nabla^2 u + V u, \tag{16}$$

with the aforementioned boundary and initial conditions. By the MLS approximation one gets $u(\mathbf{x},t) = \sum_{i=1}^{N} \Phi_i(\mathbf{x}) u_i(t) \equiv \mathbf{\Phi} \mathbf{U}_s$ for the unknown function, $u_q(\mathbf{x},t) = \sum_{i=1}^{N} \frac{\partial^2 \Phi_i(\mathbf{x})}{\partial q^2} u_i(t) \equiv \mathbf{\Phi}_{qq} \mathbf{U}_s$ for the partial *x*, *y* derivative and $u_{qq}(x,t) = \sum_{i=1}^{N} \frac{\partial^2 \Phi_i(x)}{\partial q^2} u_i(t) \equiv \mathbf{\Phi}_{qq} \mathbf{U}_s$ the second *x*, *y* partial derivative. Additionally, we set n_d as the number of nodes in the interior, n_b as the number of nodes on the boundary, and the final number of nodes as $N(N = n_d + n_b)$. The first equation, Eq. (15) can be written as

$$\frac{\partial u}{\partial t} + \left(\nabla^2 \upsilon + V \upsilon\right) = 0. \tag{17}$$

From the notation described above and using the Euler's θ -weighted time-stepping scheme for temporal discretization, for the interior nodes one gets

$$\Phi_{d} \frac{u^{n+1}-u^{n}}{\delta t} + \theta \left(\left(\Phi_{d,xx} + \Phi_{d,yy} \right) \upsilon^{n+1} + V \upsilon^{n+1} \right) + \left(1 - \theta \right) \left(\left(\Phi_{d,xx} + \Phi_{d,yy} \right) \upsilon^{n} + V \upsilon^{n} \right) = 0.$$
(18)

Multiplying both parts by δt one gets

$$\Phi_{d}u^{n+1} - \Phi_{d}u^{n} + \theta \,\delta t \left(\left(\Phi_{d,xx} + \Phi_{d,yy} \right) \upsilon^{n+1} \right) + \theta \,\delta t \left(V \upsilon^{n+1} \right) + + \left(1 - \theta \right) \,\delta t \left(\left(\Phi_{d,xx} + \Phi_{d,yy} \right) \upsilon^{n} \right) + \left(1 - \theta \right) \,\delta t \left(V \upsilon^{n} \right) = 0$$
(19)

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In matrix notation, for all points, incorporating the boundary conditions at n_b boundary nodes one has

$$\begin{bmatrix} \Phi_d & \theta \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} + V \right) \\ G_V \Phi_b & 0 \end{bmatrix} \begin{bmatrix} u^{n+1} \\ v^{n+1} \end{bmatrix}$$
$$= \begin{bmatrix} \Phi_d & -(1-\theta) \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} + V \right) \\ 0 & 0 \end{bmatrix} \begin{bmatrix} u^n \\ v^n \end{bmatrix} + \begin{bmatrix} 0 \\ g_1^{n+1} \end{bmatrix}, \quad (20)$$

where G_V is the operator defining the boundary conditions for velocity (Dirichlet 162 type on $\partial \Omega$). 163

These equations can be written in a more compact manner by setting

$$\begin{aligned} \boldsymbol{H}_{A}^{+} &= \begin{bmatrix} \Phi_{d} & \theta \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} + V \right) \\ G_{V} \Phi_{b} & 0 \end{bmatrix}, \\ \boldsymbol{H}_{A}^{-} &= \begin{bmatrix} \Phi_{d} & -(1-\theta) \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} + V \right) \\ 0 & 0 \end{bmatrix} \end{aligned}$$

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164 and $\boldsymbol{F}_A = \begin{bmatrix} 0\\ g_1^{n+1} \end{bmatrix}$,

where $\boldsymbol{H}_{A}^{+} \in \boldsymbol{R}^{N \times N}$, $\boldsymbol{H}_{A}^{-} \in \boldsymbol{R}^{N \times N}$, $\boldsymbol{\overline{\mathcal{F}}}_{A} \in \boldsymbol{R}^{N \times 1}$ and $\boldsymbol{\theta} \in \boldsymbol{\overline{R}}^{n_{d} \times 1}$.

Regarding the second Eq. (16) and following the same procedure described for Eq. (15) one can derive (in matrix notation)

$$\begin{bmatrix} \theta \delta t \left(-\Phi_{d,xx} - \Phi_{d,yy} - V\right) & \Phi_d \\ 0 & G_B \Phi_b \end{bmatrix} \begin{bmatrix} u^{n+1} \\ v^{n+1} \end{bmatrix}$$
$$= \begin{bmatrix} -(1-\theta) \delta t \left(-\Phi_{d,xx} - \Phi_{d,yy} - V\right) & \Phi_d \\ 0 & 0 \end{bmatrix} \begin{bmatrix} u^n \\ v^n \end{bmatrix} + \begin{bmatrix} 0 \\ g_2^{n+1} \end{bmatrix}, \quad (21)$$

where G_B is the operator defining the boundary conditions for the induced magnetic field on $\partial \Omega$. Once again, the above equations can be written in more compact form by setting

$$\begin{split} \boldsymbol{H}_{B}^{+} &= \begin{bmatrix} \theta \, \delta t \left(-\Phi_{d,xx} - \Phi_{d,yy} - V \right) & \Phi_{d} \\ 0 & G_{B} \Phi_{b} \end{bmatrix}, \\ \boldsymbol{H}_{B}^{-} &= \begin{bmatrix} -(1-\theta) \, \delta t \left(-\Phi_{d,xx} - \Phi_{d,yy} - V \right) & \Phi_{d} \\ 0 & 0 \end{bmatrix}, \\ \text{and} \ \boldsymbol{F}_{B} &= \begin{bmatrix} 0 \\ g_{2}^{n+1} \end{bmatrix}, \end{split}$$

where $\boldsymbol{H}_{B}^{+} \in \boldsymbol{R}^{N \times N}$, $\boldsymbol{H}_{B}^{-} \in \boldsymbol{R}^{N \times N}$ and $\boldsymbol{F}_{B} \in \boldsymbol{R}^{N \times 1}$.

The final system of the QHD coupled partial differential equations can be written as

$$\begin{bmatrix} \boldsymbol{H}_{A}^{+} \\ \boldsymbol{H}_{B}^{+} \end{bmatrix} \begin{bmatrix} u^{n+1} \\ v^{n+1} \end{bmatrix} = \begin{bmatrix} \boldsymbol{H}_{A}^{-} \\ \boldsymbol{H}_{B}^{-} \end{bmatrix} \begin{bmatrix} u^{n} \\ v^{n} \end{bmatrix} + \begin{bmatrix} \boldsymbol{F}_{A} \\ \boldsymbol{F}_{B} \end{bmatrix}.$$
(22)

Finally, setting

$$\boldsymbol{u}^n = \begin{bmatrix} u^n \\ v^n \end{bmatrix}, \quad \boldsymbol{F} = \begin{bmatrix} \boldsymbol{F}_A \\ \boldsymbol{F}_B \end{bmatrix}, \quad \boldsymbol{Q}^+ = \begin{bmatrix} \boldsymbol{H}_A^+ \\ \boldsymbol{H}_B^+ \end{bmatrix}, \quad \boldsymbol{Q}^- = \begin{bmatrix} \boldsymbol{H}_A^- \\ \boldsymbol{H}_B^- \end{bmatrix},$$

the discretized PDEs of QHD flow are summed as

$$\boldsymbol{u}^{n+1} = \left(\boldsymbol{Q}^{+}\right)^{-1} \left(\boldsymbol{Q}^{-}\boldsymbol{u}^{n} + \boldsymbol{F}\right), \qquad (23)$$

where $\boldsymbol{Q}^+ \in \boldsymbol{R}^{2N \times 2N}$, $\boldsymbol{Q}^- \in \boldsymbol{R}^{N \times 2N}$ and $\boldsymbol{F} \in \boldsymbol{R}^{2N \times 1}$.

169 4 Numerical experiments

In order to examine the validity and the effectiveness of the proposed scheme, four representative case studies were examined [Dehghan and Shokri (2007), Dehghan, and Mirzaei (2008), Dehghan, and Mirzaei (2008)]; thee cases for the linear Schrödinger equation with and without the potential function present, and a fourth one for nonlinear Schrödinger equation.

175 **Example 1**

Initially, we consider the case with potential V = 0 at the Schrödinger equation, in the spatial domain $(0,1) \times (0,1)$ and initial conditions [Dehghan, and Mirzaei (2008)]

$$\Psi(x, y, 0) = e^{i(x+y)},$$

which generates the exact solution

$$\Psi(x, y, t) = e^{i(x+y-2t)}$$

The Dirichlet boundary conditions were extracted from the analytical solution. Table 1 presents the relative error of both real and imaginary parts, defined as $\varepsilon = \frac{\|u_{num} - u_{exact}\|_2}{\|u_{exact}\|_2}$, for t = 5 and t = 20 sec. The meshless point method with MLS approximation depends on several parameters that have to be chosen properly in order to achieve convergence and accuracy. These parameters include the proper

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nodal distribution, the number of nodes in the support domain, and the user-defined 181 variables used in the weight function. For our investigation purposes we use a regu-182 lar nodal distribution of Type-I [Kim and Liu (2006)], which ensures the fulfillment 183 of the so-called positivity conditions [Jin, Li and Aluru (2004)]. Additionally, we 184 set the user-defined parameter α_0 at the weight function to be $\alpha_0 = 0.2$, the number 185 of nodes in the support domain 10, and time step dt = 0.05. As pointed out else-186 where [Bourantas, Skouras and Nikiforidis (2009)], when the number of nodes in 187 the total domain is increased, the accuracy is improved. This also is depicted at the 188 Table 1. 189

The MLS approximation is obtained by a special least-squares method [Liu and 190 Gu (2005)]. The function obtained by the MLS approximation is a smooth curve 191 (or surface), which does not pass through the nodal values inherently. Therefore, 192 the MLS shape functions do not, in general, satisfy the Kronecker delta condition. 193 Thus, when the nodes in the support domain increase, the Gaussian weight func-194 tion loses its local character (delta function property), resulting in truncated errors 195 which decrease the accuracy of the numerical results. Thus, in Table 2, we present 196 the dependence of the accuracy from the number of nodes in the support domain. 197 To do that, we used a constant grid of 31×31 nodes and altered the number of 198 nodes at the support domain. The results obtained show the very good accuracy of 199 the proposed scheme when the number of the nodes in the support domain is kept 200 small. Moreover, in Fig. 1, plots are presented for numerical and exact solutions 201 for the real and imaginary part at t = 20, using a 21×21 regular grid and 10 nodes 202 in the support domain. 203

Table 1: Relative errors at t = 5 and t = 20 for different grids, dt = 0.05 for support domain 10.

	<i>t</i> =	= 5	t = 20		
Grid	Real	Imaginary	Real	Imaginary	
11x11	7.6981E-05	1.2284E-04	1.6446E-05	2.9136E-04	
16x16	2.2556E-05	6.7048E-05	1.5732E-04	9.3423E-05	
21x21	8.7172E-06	4.8229E-05	1.1227E-04	3.6805E-05	
26x26	5.8460E-06	3.9876E-05	7.9893E-05	1.7957E-05	
31x31	5.4051E-06	3.5865E-05	5.9303E-05	1.4117E-05	

204 *Example 2*

As a second exarting we consider the Schrödinger equation in the spatial domain $(0,1) \times (0,1)$, and with potential function [Dehghan and Shokri (2007), Dehghan,

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Table 2: Relative errors at t = 5 and t = 20 for different number of nodes in the support domain, dt = 0.05.

	<i>t</i> = 5		t = 20	
Sup. Domain	Real	Imaginary	Real	Imaginary
10	5.4051E-06	3.5865E-05	5.9303E-05	1.4117E-05
15	7.1039E-05	7.1072E-05	5.9706E-05	1.4208E-05
20	6.3606E-05	1.8795E-05	2.5153E-04	7.6710E-05
25	6.2948E-04	1.9208E-03	2.2634E-03	8.1771E-04
30	5.3568E-03	7.4155E-03	6.9545E-03	2.8022E-03
35	1.5133E-02	3.0736E-02	3.0478E-02	1.2842E-02

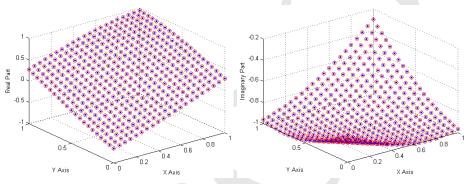


Figure 1: Plots of numerical and exact solutions for the real and imaginary part at t = 20, using a 21×21 regular grid and 10 nodes in the support domain.

and Mirzaei (2008)]

 $V(x,y) = 3 - 2\tanh^2 x - 2\tanh^2 y,$

Initial and boundary conditions are defined as

$$\Psi(x, y, 0) = \frac{i}{\cosh(x)\cosh(y)}, \quad 0 \le x, y \le 1$$

and

$$\psi(0, y, t) = \frac{ie^{it}}{\cosh(y)}, \quad \psi(1, y, t) = \frac{ie^{it}}{\cosh(1)\cosh(y)},$$
$$\psi(x, 0, t) = \frac{ie^{it}}{\cosh(x)}, \quad \psi(x, 1, t) = \frac{ie^{it}}{\cosh(x)\cosh(1)}.$$

The analytical solution is given by

$$\Psi(x,y,t) = \frac{ie^{it}}{\cosh x \cosh y}.$$

Table 3 presents the maximum absolute error for the real and the imaginary parts 205 of the solution at different times up to t = 1, using meshless point collocation 206 method with MLS approximation. For comparison purposes, numerical results are 207 also presented using meshless collocation method with global Radial Basis Func-208 tions approximation using multiquadrics (MQ) and thin plate splines (TPS) respec-209 tively [Dehghan and Shokri (2007)]. These results were obtained for dx = dy =210 0.1, and dt = 0.001. The maximum relative error, ε , defined as $\varepsilon = Max_{(x,y)\in\Omega}$ 211 $\frac{\left|u_{exact}(x,y,t) - u_{approximate}(x,y,t)\right|}{\left|u_{exact}(x,y,t)\right|}$, was also reported. The total number of nodes was 212 121 $(11 \pm \times 11)$, the number of nodes in the support domain was set to 10, ensuring 213 the inversion of the moment matrix, A(x), and the parameter α_0 was set to $\alpha_0 = 0.2$ 214 [Liu (2003)]. 215

At Table 4 the CPU time (in seconds) is presented, in order to demonstrate the efficiency of the meshless point collocation method. The shape functions are not pre-defined, and they must be constructed before the numerical solution of the resulting algebraic system. Thus, in our in-house code, the numerical procedure contains two parts; first comes the construction of the shape functions and, then, the solution of the resulting linear system.

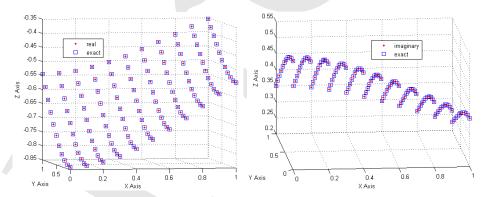


Figure 2: Plots of numerical and exact solutions for the real and imaginary part at t = 1, using a 11×11 grid.

In Fig. 2 the graphs of the real part and the imaginary parts of the numerical and the analytical solutions using MLS are shown at time t = 1, with dx = dy = 0.1,

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Table 3: Maximum absolute error of multiquadrics and thin plate spline based scheme at different times with dx = dy = 0.1, dt = 0.001 and c = 0.7 for MQ. For every value of *t*, the first and second rows of data correspond to the use of MQ, TPS as the radial basis function respectively and the third for the MPC.

~		Maximum a	bsolute error
t		Real	Imaginary
0.1	MQ	2.4407E-05	2.9974E-05
0.3	TPS	7.8895E-05	9.8635E-05
0.5	MLS	1.4644E-04	1.5220E-04
0.7	MQ	2.9466E-05	2.3861E-05
1.0	TPS	1.0368E-05	8.6876E-05
	MLS	1.3317E-04	9.8297E-05
	MQ	2.7468E-05	3.4044E-05
	TPS	7.7545E-05	9.1676E-05
	MLS	8.3716E-05	1.9683E-04
	MQ	2.5495E-05	1.8694E-05
	TPS	8.9137E-05	7.7454E-05
	MLS	1.5182E-05	1.7088E-05
	MQ	2.9444E-05	2.4222E-05
	TPS	1.0626E-04	9.3425E-05
	MLS	1.5138E-04	9.5315E-05

Table 4: CPU time in seconds for shape construction and solution of the resulting transient, linear system.

nodes	Shape Functions (sec)	Linear system (sec)
121	0.53125	9.54687
441	1.90625	35.98437
961	3.78125	108.01562
1681	6.9375	282.06250

15

dt = 0.001. Note that there is no essential divergence between the exact solution and the numerical solution in Fig. 2, for the given accuracy.

One can notice that, for coarse grids, as in the case of 121 nodes, the numerical re-226 sults obtained by the meshless point collocation with MLS approximation are less 227 accurate than those obtained by the global multiquadrics Radial Basis Function. 228 Although full-domain RBF methods are highly flexible and can exhibit high-order 229 convergence rates [Madych and Nelson (1990)], in their basic implementation the 230 fully-populated matrix systems produced lead to poor numerical conditioning as 231 the size of the dataset increases. This problem is described by Schaback [Schaback 232 (1993)] as the "uncertainty relation", in which better conditioning is associated 233 with worse accuracy, and worse conditioning is associated with improved accuracy. 234 With increasingly large datasets and increasingly flat basis functions, this problem 235 becomes more pronounced. Thus, global RBF are not appropriate for real world 236 applications, were the number of the degrees of freedom (nodes) are large. On 237 the other hand, MLS approximation, being a localized-type approximation, uses a 238 small number of neighboring nodes for interpolation. This makes the MLS approx-239 imation more suitable for many applications arising in science and engineering. 240 Furthermore, the small number of nodes used makes the method computationally 241 time and memory saving. This is evident at Table 5 where doubling the nodal dis-242 tribution density increases the accuracy of the numerical solution by an order of 243 magnitude, while the computational efficiency of the scheme is retained. 244

$\overline{\mathbf{r}}$	Maximum absolute error		Maximum relative error		
t		Real	Imaginary	Real	Imaginary
0.1	dx = 0.05	3.6969E-05	3.6482E-05	1.6920E-04	1.9557E-05
0.3	dx = 0.025	8.6811E-06	9.0573E-06	4.3579E-05	4.9694E-06
0.5	dx = 0.05	3.9436E-05	2.9388E-05	6.2200E-05	0.4472E-05
0.7	dx = 0.025	9.4746E-06	7.6511E-06	1.6000E-05	3.8843E-06
1.0	dx = 0.05	2.3815E-05	4.1528E-05	2.8327E-05	2.6131E-05
	dx = 0.025	7.6615E-06	1.0489E-05	7.3170E-06	6.4270E-06
	dx = 0.05	4.2027E-05	1.8415E-05	3.4707E-05	1.2463E-05
	dx = 0.025	9.9946E-06	4.7105E-06	9.0405E-06	3.8419E-06
	dx = 0.05	2.5305E-05	3.4800E-05	1.7559E-05	3.8282E-05
	dx = 0.025	6.1169E-06	9.7405E-06	3.6556E-06	9.9219E-06

Table 5: Absolute and relative errors at different times for dx = dy = 0.05 and 0.025, and dt = 0.001.

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245 *Example 3*

Following, we consider the Schrödinger equation in $(0,1) \times (0,1)$ spatial domain and with potential function [Dehghan and Shokri (2007), Dehghan, and Mirzaei (2008)]

$$V(x,y) = 1 - \frac{2}{x^2} - \frac{2}{y^2}$$

and initial and boundary conditions

$$\psi(x, y, 0) = x^2 y^2$$

and

$$\Psi(0, y, t) = 0, \quad \Psi(1, y, t) = y^2 e^{it}, \quad \Psi(x, 0, t) = 0, \quad \Psi(x, 1, t) = x^2 e^{it},$$

The analytical solution is given as

 $\psi(x, y, t) = x^2 y^2 e^{it}.$

Table 6 presents the maximum absolute error for the real part and imaginary part at different times up to t = 1, using MLS approximation and time step dt = 0.05. The results obtained were compared with those obtained using the multiquadrics and the plate spline RBF with the same nodal distribution and time step, dt = 0.0005 [Dehghan and Shokri (2007)]. One can observe that, for MPC with MLS approximation of localized type, using a time step two orders lower than the time step used in global RBF, the absolute errors present two orders higher accuracy.

Finally, in Fig. 3, the graphs of the real part and the imaginary parts of the numerical and the analytical solutions using MLS are shown at time t = 1, with dx = dy = 0.1, dt = 0.05. Note that there is no essential divergence between the exact solution and the numerical solution in Fig. 2, for the given accuracy.

257 Example 4

Finally, we consider the generalized nonlinear two-dimensional Schrödinger equation written as [Dehghan, and Mirzaei (2008)]:

$$-i\frac{\partial\psi}{\partial t} + \frac{\partial^{2}\psi}{\partial x^{2}} + \frac{\partial^{2}\psi}{\partial y^{2}} = B(x, y, t) \psi + C(x, y, t) |\psi|^{p} \psi,$$

with the initial and boundary conditions

 $\psi(x, y, 0) = \cos(x)\cos(y), \quad (x, y) \in \Omega$

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Table 6: Maximum absolute error of multiquadrics and thin plate spline-based scheme at different times with dx = dy = 0.1, dt = 0.0005 and c = 0.45 for MQ. For every value of *t*, the first and second rows of data correspond to the use of MQ and TPS as the radial basis function, respectively and the third for the MPC when dx = dy = 0.1, dt = 0.05.

	Maximum absolute error		
	Real	Imaginary	
MQ	4.0410E-04	3.5722E-04	
TPS	8.6297E-04	8.3522E-04	
MLS	2.7156E-06	1.1912E-06	
MQ	5.1291E-04	3.0509E-04	
TPS	8.0754E-04	7.1756e-04	
MLS	3.1253E-06	1.5355E-06	
MQ	4.6396E-04	3.9520E-04	
TPS	5.0822E-04	7.7982E-04	
MLS	1.7575E-06	2.2252E-06	
MQ	3.8999E-04	4.1646E-04	
TPS	7.5356E-04	9.2228E-04	
MLS	2.2781E-06	3.7907E-06	
MQ	3.7209E-04	4.1267E-04	
TPS	6.5917E-04	8.9195E-04	
MLS	1.4423E-06	1.0944E-06	
	TPS MLS MQ TPS MLS MQ TPS MLS MQ TPS MLS MQ TPS	RealMQ4.0410E-04TPS8.6297E-04MLS2.7156E-06MQ5.1291E-04TPS8.0754E-04MLS3.1253E-06MQ4.6396E-04TPS5.0822E-04MLS1.7575E-06MQ3.8999E-04TPS7.5356E-04MLS2.2781E-06MQ3.7209E-04TPS6.5917E-04	

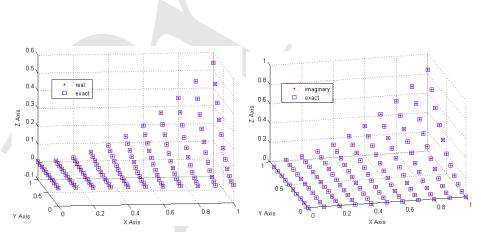


Figure 3: Plots of the exact and the numerical solution at t = 1.0.

and Neumann boundary conditions on all sides of the spatial domain

$$\frac{\partial \psi}{\partial \boldsymbol{n}} = 0$$

The analytical solution is given as

 $\Psi(x, y, t) = e^{-it} \cos(\pi x) \cos(\pi y).$

The lagging of coefficients method has been utilized to eliminate the non-linearity of the examined problem. The spatial domain of the problem is defined as $0 \le x, y \le 1$. The function used in the present problem are defined as $C(x, y) = 1 - 2\pi^2$, $B(x, y) = (1 - 2\pi^2)(1 - \cos^2(\pi x)\cos^2(\pi y))$ and p = 2. We have to notice that the accuracy of the case under consideration agrees with the exact solution at about two significant digits and, as the time increases it becomes worse. This is due to the imposition of the Neumann boundary conditions. When using Dirichlet boundary conditions the accuracy of the numerical results increases. Following the aforementioned procedure the final linearized system in matrix notation can be written as

$$\begin{split} \boldsymbol{H}_{A}^{+} &= \begin{bmatrix} \Phi_{d} & \theta \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} - B \Phi_{d} - C \left(|\Psi|^{p} \right)^{n} \Phi_{d} \right) \\ G_{V} \Phi_{b} & 0 \end{bmatrix}, \\ \boldsymbol{H}_{A}^{-} &= \begin{bmatrix} \Phi_{d} & -(1-\theta) \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} - B \Phi_{d} - C \left(|\Psi|^{p} \right)^{n} \Phi_{d} \right) \\ 0 & 0 \end{bmatrix} \\ \text{and} \ \boldsymbol{F}_{A} &= \begin{bmatrix} 0 \\ g_{1}^{n+1} \end{bmatrix}, \\ \boldsymbol{H}_{B}^{+} &= \begin{bmatrix} \theta \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} - B \Phi_{d} - C \left(|\Psi|^{p} \right)^{n} \Phi_{d} \right) \quad \Phi_{d} \\ 0 & G_{B} \Phi_{b} \end{bmatrix}, \\ \boldsymbol{H}_{B}^{-} &= \begin{bmatrix} -(1-\theta) \, \delta t \left(\Phi_{d,xx} + \Phi_{d,yy} - B \Phi_{d} - C \left(|\Psi|^{p} \right)^{n} \Phi_{d} \right) \quad \Phi_{d} \\ 0 & 0 \end{bmatrix}, \\ \text{and} \ \boldsymbol{F}_{B} &= \begin{bmatrix} 0 \\ g_{2}^{n+1} \end{bmatrix}. \end{split}$$

259 5 Conclusions

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In the present work we used the meshless numerical scheme to solve the twodimensional time-dependent linear and nonlinear Schrödinger equation using the point collocation method with MLS approximation. For the Schrödinger equation we developed a fully coupled, transient, and strong-form solver for the real and

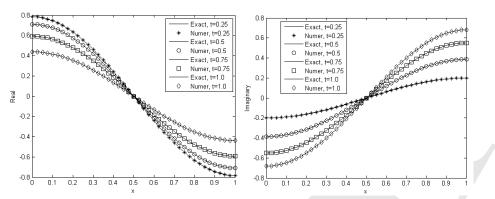


Figure 4: Analytical and numerical solutions at various time.

the imaginary parts of the general solution of the so-called quantum hydrodynamic 264 (QHD) equation. The proposed scheme is applied to four benchmark cases hav-265 ing analytical solutions, with our results exhibiting excellent agreement with all the 266 analytical ones. The numerical results were also compared with those provided by 267 another collocation method, that is, the global Radial Basis Function method. The 268 numerical results provided by the proposed scheme are highly accurate, compared 269 with the ones provided by the multiquadrics and the thin plates splines RBF. Fur-270 thermore, in some cases they are also less CPU time and memory consuming. This 271 makes the application of the MLS approximation very attractive for the numerical 272 solution of this kind of physical problems. 273

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